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Review

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Features of synthesis and properties of new materials based on monoisotopic silicon and germanium. Review

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Abstract

This paper reviews scientific works on the preparation and properties of isotopically enriched silicon and germanium, along with their compounds. It covers the technological aspects and peculiarities of synthetic methods with the deep purification processes for obtaining isotopically enriched silicon and germanium compounds. The review also discusses the production of polycrystalline and single-crystal samples with varying degrees of isotopic and chemical purity. The results of a study investigating the physicochemical characteristics of both simple and complex substances derived from isotopically enriched silicon and germanium are presented. These studies indicate that the isotopic composition of silicon and germanium significantly affects heat capacity, thermal conductivity, and light absorption processes. Finally, the paper explores current applications of substances and materials based on isotopically enriched silicon and germanium.

Keywords: Silicon, Germanium, Isotopes, Hydrides, Materials

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Conventionally, the characteristics that determine the set of physical and chemical properties of substances and materials include morphology, the level of chemical purity, and structural perfection. However, as a rule, the presence of several isotopes in the natural composition of most chemical elements is often ignored. Yet, these isotopes can be considered as a combination of the main and impurity isotopes of the chemical element [1, 2].

Differences in the properties of isotopically modified substances (the so-called isotopic effect) are due to the difference in the molar masses of isotopes. What is more, they are more noticeable in the chemical elements in periods 1–2 of the Mendeleev periodic table [3]. A detailed study and expansion of the database of properties of individual isotopes of chemical elements in the form of simple substances and compounds is important for fundamental and applied science. This information is relevant for predicting the properties and for the manufacturing of isotopic compositions, i.e. mixtures with desired properties, as part of isotopic engineering [4].

In the case of chemical elements with a relatively small difference in the mass numbers of isotopes, it is necessary to obtain corresponding substances with high chemical purity and structural perfection which will allow a definitive determination of the isotopic effects. High-tech semiconductor materials, monocrystalline silicon and germanium, are the most convenient for the study of isotopic effects. They are currently considered as the most advanced materials in terms of structural perfection and chemical purity.

Silicon is represented in nature by three stable isotopes: 28 Si (92.230%), 29 Si (4.683%), and 30 Si (3.087%). Among the radioactive isotopes of silicon, 32 Si ($T_{1/2} \sim 150$ years) is the longest-lasting [3]. Germanium consists of four stable isotopes, 70 Ge (20.84%), 72 Ge (27.54%), 73 Ge (7.73%), 74 Ge (36.28%), and 76 Ge (7.61%), with a half-life of about $2 \cdot 10^{21}$ years [3]. Gas ultracentrifugation of volatile fluorides is mainly used to separate the isotopes of these chemical elements [5–7]; monogermane is also used in the separation of germanium isotopes [8, 9].

The first works dedicated to the production and properties of monoisotopic silicon and germanium were published in the 1990s. In paper [10] the Czochralski technique with a natural silicon seed was used to produce a single crystal of silicon-28 with a weight of about 300 g and an isotope content of ²⁸Si at 99 mol %. This single crystal was used to measure the parameter of the silicon-28 lattice and to determine the Avogadro's number. The initial silicon-28 was obtained by aluminum reduction of silicon dioxide-28 with a 99.88% silicon-28 enrichment. The content of oxygen, carbon, and boron impurities in silicon-28 was 10¹⁶ - 10¹⁸ atoms/cm³. The authors of [11] obtained a single crystal of silicon-28 with a diameter of 4 mm and a length of 50 mm and a 99.924 at. % isotope content of ²⁸Si. The silicon powder obtained by the aluminum reduction of silicon oxide-28 was compressed, sintered, and zone-melted. The resulting sample had a p-type conductivity and contained an aluminum impurity at 1.10^{-3} at. %. According to the authors, isotopically enriched silicon was a promising material for the development of a quantum computer.

In [12], the liquid-phase epitaxy was used to obtain a film of silicon-28 with a 99.7% enrichment and a thickness of 11.7 microns from the indium melt. The film was produced on a natural silicon substrate, on which an aluminum layer with a thickness of about 1000 Å was sprayed. After measuring the thermal conductivity of the sample, it was found that the thermal conductivity of isotopically enriched ²⁸Si at 298 K was about 60% higher than that of silicon with a natural isotopic composition. However, as a result of interlaboratory measurements using the technique of stationary heat flux, it was found that the difference in the thermal conductivity of isotopically enriched ²⁸Si and natural silicon did not exceed 10±2% [13]. The significant discrepancy in the value of the thermal conductivity of silicon-28 obtained in [12] could be due to the contamination of the sample.

The researchers also obtained isotopically enriched samples of ²⁹Si and ³⁰Si in the form of single crystals [5,14] and epitaxial films [15] with the degree of enrichment in the main isotope of silicon that did not exceed 99.9%. In [5], polycrystalline silicon-29 and silicon-30 were produced by the thermal decomposition of monosilane which had been obtained by the reaction between silicon tetrafluoride (²⁹SiF₄,

³⁰SiF₄) and calcium hydride. Monoisotopic silicon was deposited on a molybdenum wire, resulting in granules with a size of 0.3 – 0.5 mm. To minimize isotopic dilution at the stage of growing single crystals by the Czochralski technique, smallvolume crucibles with a coating of 29SiO2 or ³⁰SiO₂ with a thickness of 100-150 microns were used. The content of the main isotope of silicon in the obtained ²⁹Si and ³⁰Si single crystals was 99.225±0.023 at. % and 99.742±0.015 at. %; in the initial ²⁹SiF₄ and ³⁰SiF₄, the content of ²⁹Si, ³⁰Si in the composition of silicon was 99.576±0.105 at. % and 99.829 ± 0.060 at. %. In [14], silicon-29 was precipitated on a graphite substrate, after the removal of which the resulting silicon-29 was melted into a rod. A natural silicon seed was used to produce a single crystal of silicon-29.

The authors of [16] measured the temperature dependence of the thermal conductivity of a single crystal of isotopically enriched silicon-29 (99.919%) in the range of 2.4 - 410 K. It was found that at low temperatures (T < 6 K) in the mode of boundary phonone scattering, the thermal conductivity of the ²⁹Si crystal was higher than that of the ²⁸Si crystal (99.983%). In the high temperature region, where thermal conductivity is determined by anharmonic processes of phonon scattering, the value of thermal conductivity of ²⁹Si was lower than that of ²⁸Si.

To obtain significant quantities of silicon-28, as well as samples of silicon-29 and silicon-30, the Institute of Chemistry of High Purity Substances of the Russian Academy of Sciences developed a laboratory technology for the synthesis, deep purification, and thermal decomposition of monosilane for the production of polycrystalline silicon-28 [1,2]. Further studies aimed at obtaining and improving the isotopic and chemical purity were triggered by using monoisotopic silicon-28 to determine the Avogadro's number and creating a physically justified kilogram standard within the framework of the Avogadro and Kilogram projects [6, 7, 10, 17, 18]. Silicon tetrafluoride SiF, was used as the starting material. It was enriched by gas ultracentrifugation (ZAO Tsentrotekh-SPb, AO Production Association "Electrochemical Plant") to the content of silicon-28 of over 99.99% [7]. The content of molecular impurities in samples of isotopically enriched ${}^{28}{\rm SiF}_{_4}$ according to high-resolution Fourier IR spectroscopy [19] and gas chromatography [20] was: $Si_2F_6O - 2\cdot 10^{-1}$ mol %; $CO_2 - 1\cdot 10^{-2}$ mol %; $H_2O - 6\cdot 10^{-3}$ mol %; $HF - 3\cdot 10^{-4}$ mol %; hydrocarbons C_1-C_4 (methane, ethane, ethylene, propane, butane) $-n\cdot 10^{-3}-n\cdot 10^{-5}$ mol % [21]. The content of metal impurities in isotopically enriched silicon tetrafluoride, determined by the atomic emission with matrix distillation for the concentration of non-volatile impurities, was at $n\cdot 10^{-7}$ wt. % [22, 23]. The content of silicon isotopes in the composition of silicon tetrafluoride, as well as monosilane and crystalline silicon, was determined by mass spectrometry [24].

 $^{28}\mbox{SiH}_{\mbox{\tiny 4}}\mbox{was synthesized by the reaction of}$ ²⁸SiF₄ with ground calcium hydride in a hydrogen flow reactor. Calcium hydride was synthesized from distilled calcium and hydrogen [25] and ground before the synthesis (in situ); calcium hydride was ground to a particle size of less than 1 mm and was loaded into the monosilane synthesis reactor in boxes with an inert nitrogen atmosphere. This ensured a low content of oxygen-containing impurities in calcium hydride, which are formed upon contact with atmospheric moisture, and a high isolated yield of monosilane. The productivity of the laboratory synthesis was 3 kg of ²⁸SiH₄ per month. However, this technique for monosilane synthesis has a disadvantage, i.e. the low degree of conversion of the solidphase reagent, CaH2, which does not exceed 15% [26]. Therefore, it is important to increase the degree of conversion of calcium hydride. One of the solutions can be to carry out the conversion process at an elevated pressure [27] or to use mechanical activation, i.e. grinding the solid-phase reagent in the process of producing monosilane in a rotating flow reactor [28].

The isolated yield of $^{28}SiH_4$ over $^{28}SiF_4$ was 92–96%. According to [26], the obtained monosilane contains impurities of hydrocarbons, siloxanes, and polysilanes at $n \cdot 10^{-3} - n \cdot 10^{-5}$ mol %. One of the sources of hydrocarbon impurities in monosilane is the hydrogenation of impurity carbon in calcium during the contact of calcium hydride with hydrogen at an elevated temperature [21]. Preliminary purging of the reactor with calcium hydride with high-purity hydrogen at 200 °C allows reducing the content of unsaturated hydrocarbons in the resulting silane and disiloxane impurities by about an order of magnitude [29].

A combination of cryofiltration and lowtemperature rectification was used for deep purification of isotopically enriched monosilane [30]. Cryofiltration on a Petryanov filter at a temperature of about 165 K allowed separating the higher-boiling impurities of siloxanes and higher silanes (disilane, trisilane). Rectification was carried out in metal rectification columns with a middle feeding tank in the mode of discrete choice of impurities from the lower and upper sections filled with a nichrome wire spiralprismatic nozzle [29]. The yield of the high-purity product at the purification stages was 80-90%. The content of chemical impurities in samples of high-purity monosilane was: alkylsilanes at $n\cdot 10^{-6}$ mol %, hydrocarbons C_1-C_9 at $n\cdot 10^{-5}$ mol %, disiloxane at 10⁻³ mol %, and higher silanes at $n \cdot 10^{-2}$ mol %.

Monoisotopic polycrystalline silicon (28Si, 29Si, ³⁰Si) was obtained by the pyrolysis of high-purity isotopically enriched monosilane in SAA-20 and SAA-800 monosilane decomposition plants (Steremat Electrowarme), which differed in the maximum amount of obtained material (150 g and 6 kg, respectively). In the absence of rods from isotopically enriched material and due to the restrictions regarding the usage of silicon rods of natural isotopic composition to avoid isotopic dilution, monoisotopic silicon was deposited on molybdenum substrates. This involved obtaining silicon polycrystals, which were connected into a tape by means of contactless high-frequency welding. The resulting tape was used as a substrate for silicon precipitation [31]. Thin walled stainless steel tubes (12X18H10T) can also be used as a substrate for precipitation. After precipitation, the substrate material dissolved in hydrochloric acid and the resulting silicon tube was melted into a rod of the specified diameter by crucibleless melting. This technique is simpler and reduces the cost for the production of seed rods, which is especially important when obtaining samples of monoisotopic silicon of various nature and variable isotopic composition.

The manufacture of monocrystalline silicon samples involved using crucibleless melting and the Czochralski technique. To avoid isotopic dilution when using the Czochralski technique with quartz crucibles, special techniques were developed for applying protective coatings from isotopically enriched silicon dioxide by oxidizing monosilane ²⁸SiH₄ [32].

One of the results of the *Avogadro* international project was the production of two spheres of monocrystalline silicon-28 with a diameter of 96.3 mm and a mass of 1 kg. Measuring its volume, density, atomic lattice constant, and molar mass allowed obtaining the Avogadro constant, $N_{\rm A} = 6.022\ 14082(18)\cdot10^{23}\ {\rm mol}^{-1}$ with a relative error of $2\cdot10^{-8}$ [7].

The properties of isotopically enriched silicon samples are presented in [13,16,33-41] and generalized in [2]. The difference in the properties of samples of monoisotopic silicon and silicon of the natural isotopic composition is associated with a different value of the average atomic mass and with an uneven distribution of isotopes in the sample. A number of properties dependent on spin interactions (nuclear magnetic resonance spectra, electron paramagnetic resonance, and the parameters of spin-spin and spin-lattice interactions) are affected by the concentration of silicon-29 with a non-zero nuclear spin. Table 1 shows the properties of monocrystalline samples of ²⁸Si, ²⁹Si, and ³⁰Si and pure silicon of the natural isotopic composition natSi [2].

Samples of isotopically enriched germanium were obtained using a hydride technology [42]. Unlike the production of silicon isotopes, germanium did not require converting the working gas into volatile hydride for ultracentrifugation. Monogermane was used to separate germanium isotopes [8]. The technique for obtaining high-purity germanium isotopes included the stages of low-temperature rectification and thermal decomposition of monogermane and additional purification of germanium by zone melting [9]. Rectification purification of monogermanes was carried out on a metal column with a middle cube, equipped with an automated refrigerant supply system and a gas chromatograph for operational control of the impurity composition of the fractions. The weight of the monogermane loaded into the column was about 250 g, the duration of purification was about 40 hours, the isolated yield of the rectified product was 75-80%. The content of molecular impurities (C₁-C₀ hydrocarbons and their halogen derivatives, alkyl and halogen germanes, etc.) in purified monogermane samples according to chromatography-mass spectrometry [43] was at $n \cdot 10^{-5} - n \cdot 10^{-6}$ mol %.

Table 1. Properties of samples of ²⁸Si, ²⁹Si, ³⁰Si and high-purity silicon of natural isotopic composition ^{nat}Si [2]

		The content			
Property	²⁸ Si	29.0:	30 C :	natC:	of the main
	2051	²⁹ Si	³⁰ Si	natSi	isotope, %
Density at 20 °C, g·cm ⁻³	2.320071	_	-	2.329045	²⁸ Si – 99.99
					²⁸ Si – 99.98
Debye temperature, K	641	627	616	638	²⁹ Si – 99.0
					³⁰ Si – 99.0
					²⁸ Si – 99,98
Heat capacity at 100 K, J·mol⁻¹·K⁻¹	7.24	7.40	7.61	7.24	²⁹ Si – 99,0
					³⁰ Si – 99,0
Thormal conductivity at 25 V W am W	200	214		45.6	²⁸ Si – 99.98
Thermal conductivity at 25 K, W⋅cm¹⋅K¹	288	214	_	45.6	²⁹ Si – 99.92
					²⁸ Si - 99.993
Raman shift at 20 °C, cm ⁻¹	521.4	512.3	503.9	520.8	²⁹ Si – 99.92
					³⁰ Si – 99.97
					²⁸ Si - 99.993
Refractive index at 293 K, λ =1.5 microns	3.48207	3.48171	3.48112	3.48191	²⁹ Si – 99.92
•					³⁰ Si – 99.97

Thermal decomposition of monogermane was performed at a temperature of 500 °C in a quartz tube with a pyrocarbon coating. As a result of the monogermane decomposition, polycrystalline and powdered germanium was formed in the reactor, which was then melted into an ingot; the isolated yield of germanium was about 95%. Additional purification of monoisotopic germanium was carried out by zone melting in quartz boats coated with amorphous silicon dioxide in the pure hydrogen atmosphere. Germanium single crystals were grown by the Bridgman technique and the Czochralski technique [2, 5–7].

When crystals of isotopically enriched germanium are grown by the Czochralski technique, there is a problem associated with the lack of oriented isotopically enriched monocrystalline seeds. Growing single crystals of isotopically enriched germanium using a germanium seed of a natural isotopic composition leads to isotopic dilution, which can reach $n \cdot 10^{-1}$ %. To reduce the isotopic dilution of germanium, crystals were grown stage by stage: the first stage involved growing in a quartz crucible a single crystal with a diameter of 5 mm and a length of 50 mm on a seed from germanium of a natural isotopic composition using a small amount of melt of isotopically enriched germanium. This single crystal was then used as a seed for growing a single crystal of isotopically enriched

germanium. It is also possible to use a seed from previously obtained isotopically enriched germanium to minimize a certain germanium isotope. For example, when obtaining a single crystal of germanium-70, germanium-74 was used as a seed [42].

The isotopic composition of germanium in samples of isotopically enriched monogermane and germanium was determined by mass spectrometry [44]. To date, the Institute of Chemistry of High Purity Substances of the Russian Academy of Sciences have obtained and described all stable isotopes of germanium with a content of the main substance of over 99% [42, 45]. Table 2 shows the isotopic composition of germanium in high-purity monogermanes and isotopically enriched germanium [2, 42].

Some properties of the samples of isotopically enriched germanium and its hydrides were studied in [46–53].

In [46], differential scanning calorimetry was used to determine the melting point for the samples of germanium isotopes ⁷²Ge, ⁷³Ge, ⁷⁴Ge, and ⁷⁶Ge with the content of the main isotope of 99.98439%, 99.8995%, 99.9365%, and 88.21%, respectively. It was established that the melting point of samples of isotopically enriched germanium decreased with an increase in the atomic mass of the isotope. The change was 0.15 K per unit of atomic mass, which is

Table 2. The content of germanium isotopes in the composition of the obtained samples of high-purity monogermans and isotopically enriched germanium

Sub-	The content of the germanium isotope, at. %						
stance	⁷⁰ Ge	⁷² Ge	⁷³ Ge	⁷⁴ Ge	⁷⁶ Ge		
⁷⁰ GeH₄	99.99303±0.00012	0.00678±0.00009	0.00005±0.00003	0.00012±0.00005	<0.0001		
⁷⁰ Ge	99.99329±0.00006	0.00669±0.00005	0.00002±0.00001	0.00001	0.00001		
$^{72}\mathrm{GeH}_{4}$	0.005±0.001	99.953±0.008	0.018±0.003	0.023±0.006	0.002±0.001		
⁷² Ge	0.00009±0.00002	99.98439±0.00091	0.01191±0.00073	0.00356±0.00030	0.00005±0.00001		
$^{73}\mathrm{GeH}_{4}$	0.0012±0.0005	0.0415±0.0080	99.8942±0.0100	0.0626±0.0100	0.0005±0.0003		
⁷³ Ge	0.0001±0.0001	0.0391±0.0027	99.8995±0.0155	0.0611±0.0102	0.0002±0.0001		
⁷⁴ GeH ₄	0.0007±0.0003	0.0017±0.0004	0.059±0.003	99.9377±0.0031	0.0033±0.0006		
⁷⁴ Ge	0.0001±0.0001	0.0009±0.0004	0.0595±0.0010	99.9365±0.0011	0.0030±0.0003		
⁷⁶ GeH₄	0.12±0.01	0.17±0.01	0.07±0.01	11.50±0.50	88.14±0.55		
⁷⁶ Ge	0.060±0.009	0.090±0.009	0.051±0.009	11.592±0.192	88.207±0.188		

qualitatively consistent with the value calculated by the Lindemann's formula, taking into account the effect of "isotopic compression" of the unit cell. The dependence of the melting point $T_{\rm m}(^{\circ}{\rm C})$ on the average atomic mass M can be described by the equation $T_{\rm m}=949.3-0.153\cdot{\rm M}$.

In [47], differential scanning calorimetry was used to remeasure the melting point of the samples studied in [46] ⁷²Ge, ⁷³Ge, ⁷⁴Ge, and ⁷⁶Ge and to measure the melting point of the sample of germanium-70 with a content of the main isotope of 99.99329%. The melting point for samples ⁷⁰Ge, ⁷²Ge, ⁷³Ge, ⁷⁴Ge, and ⁷⁶Ge was 1,212.0 K, 1,211.6 K, 1,211.3 K, 1,211.1 K, and 1,210.8 K, respectively.

The temperature dependence of the thermal conductivity of single crystals of isotopically enriched germanium ⁷⁰Ge (degree of enrichment 99.926%), ⁷²Ge (99.980%), ⁷⁴Ge (99.921%) in the temperature range of 80–310 K was measured in [48]. In the studied temperature range, the thermal conductivity of germanium was determined by anharmonic processes of phonon scattering. A decrease in the value of thermal conductivity was also experimentally observed with an increase in the mass of the germanium isotope.

Optical spectroscopy was used to study the optical transmission of the germanium isotopes 70 Ge, 72 Ge, 73 Ge, 74 Ge, and 76 Ge in the terahertz spectral range of up to 3,000 µm [49]. It was shown that the minimum absorption within the wavelength range of 30-3,000 µm was within the range of 200-800 µm, and the corresponding

absorption coefficient for this range was less than 1 cm $^{-1}$ for most of the studied monocrystalline samples. Within the range of 1,000–3,000 μ m, there was a trend towards an increase in the absorption coefficient with an increase in the mass number of the germanium isotope.

Paper [50] presents the results of the precision measurement of the refractive index of stable single crystals of germanium isotopes ⁷²Ge, ⁷³Ge, ⁷⁴Ge, and ⁷⁶Ge with high enrichment by Fourier transform interference refractometry in the range of 1.94–20 µm with a resolution of 0.1 cm⁻¹. This paper provides coefficients of the generalized Cauchy dispersion function, approximating experimental values for the refractive index over the entire measurement range, and the transmission and Raman light scattering spectra.

The authors of [51] conducted a single experiment to investigate the dependence of the position of phonon absorption peaks in the range of 11–40 µm for five single-crystal samples of germanium, ⁷⁰Ge, ⁷²Ge, ⁷³Ge, ⁷⁴Ge, and ⁷⁶Ge. The obtained dependence can be used to identify the nature of single crystals of isotopically enriched germanium [52].

The authors of [53] calculated the standard thermodynamic functions of isotopically enriched monogermanes, ⁷⁰GeH₄, ⁷²GeH₄, ⁷³GeH₄, ⁷⁴GeH₄, and ⁷⁶GeH₄, in the temperature range of 200–700 K in the approximations "harmonic generator – rigid rotator", "anharmonic generator – oscillating non-rigid rotator", as well as by the direct

summation of experimental energy values. They established the influence of the isotopic effect on the values of the standard isobaric heat capacity, entropy, enthalpy of heating and the reduced isobaric-isothermal potential of germanium. The authors formulated the limiting requirements for the accuracy of spectral parameters used to identify the influence of the isotopic effect on thermodynamic functions and interatomic distances in the germane molecule.

In addition to fundamental and metrological studies, isotopically enriched silicon and germanium, monosilane and monogermane were investigated in the works dedicated to creating prototypes of quantum devices. State and private programs for the development and application of quantum devices are run in many countries all over the world [54]. In Russia, the systematic development of quantum technologies is carried out according to the Roadmap for the Development of the High-Tech Area "Quantum Computing" [55], which provides for various approaches to the implementation of quantum devices. One of the most promising areas is the model of a solid-state quantum computer using the state of the nuclear spin of an atom as a carrier of quantum information, a qubit [56]. An ensemble of qubits is created by implanting atoms of elements with a non-zero nuclear spin into a crystalline matrix. In the case of silicon, these are atoms of silicon-29 (nuclear spin 1/2+) distributed in a crystalline matrix of silicon-28 atoms, while in the case of germanium, these are atoms of germanium-73 (nuclear spin 9/2+) in a matrix of atoms of a common isotope (⁷⁰Ge, ⁷²Ge, ⁷⁴Ge). Of interest are heterostructures based on silicon

and germanium isotopes [57], for the creation of which molecular beam epitaxy and vapor deposition methods are used. The precursors for the production of epitaxial structures by molecular beam epitaxy are elementary silicon and germanium, whereas their volatile hydrides are the precursors for chemical vapor deposition (monosilane and monogermane).

To establish the influence of the degree of isotopic enrichment on the performance of silicon-based qubits in order to determine the limiting, physically justified values of silicon and germanium enrichment for the creation of qubits, we need isotopically enriched substances (silicon, germanium, monosilane, and monogermane) with different isotope content of ²⁹Si, ⁷³Ge. Samples of these isotopically enriched substances were prepared according to the techniques described above. Isotopically enriched silicon tetrafluoride with the content of the main isotope ²⁸Si at 99.9%, 99.99%, 99.99% and monogermane with the content of the main isotope ⁷²Ge at 99.9% were used as starting substances [58].

Table 3 shows the content of silicon isotopes in the composition of samples of ²⁸SiF₄, ²⁸SiH₄, ²⁸Si with different levels of enrichment.

From Table 3, it can be concluded that there was a statistically significant dilution of sample N°3 with an isotopic enrichment of 99.999% at the stage of silicon-28 tetrafluoride conversion to monosilane, which may be due to the intake of ²⁹Si, ³⁰Si from the material of the synthesis equipment or storage tanks.

Table 4 provides information on the content of impurities of a number of substances (hydrocarbons, halogen derivatives

Table 3. The content of silicon isotopes in samples of isotopically enriched silicon tetrafluoride-28 with different levels of enrichment

Sample		The content of the silicon isotope, wt. %					
Salli	pie	²⁸ Si	²⁹ Si	³⁰ Si			
	²⁸ SiF ₄	99.96034±0.00160	0.03957±0.00160	0.00009±0.00006			
Nº1 (3N)	²⁸ SiH ₄	99.9668±0.0015	0.0329±0.0015	0.0003±0.0001			
²⁸ Si		99.9657±0.0015	0.0340±0.0008	0.0003±0.0001			
	²⁸ SiF ₄	99.99623±0.00022	0.00369±0.00022	0.00008±0.00002			
Nº2 (4N)	²⁸ SiH ₄	99.99577±0.00028	0.00418±0.00028	0.00005±0.00001			
	²⁸ Si	99.99581±0.00045	0.00411±0.00043	0.00008±0.00002			
	²⁸ SiF ₄	99.99917±0.00011	0.00080±0.00011	0.00003±0.00001			
Nº3 (5N)	²⁸ SiH ₄	99.99882±0.00010	0.00112±0.00010	0.00006±0.00001			
	²⁸ Si	99.99881±0.00010	0.00114±0.00010	0.00005±0.00001			

Table 4. The content of molecular impurities in high-purity ²⁸SiH₄

Impurity	Content, % mol.	Impurity	Content, % mol.
Ar	(4.3±0.6)·10 ⁻⁵	trans-1,2-C ₂ H ₂ F ₂	<5·10 ⁻⁷
CO_2	<2.10-6	cis-1,2-C ₂ H ₂ F ₂	<4·10 ⁻⁷
CH_4	<5⋅10⁻⁶	1,1,1,2-C ₂ H ₂ F ₄	<5·10 ⁻⁷
C_2H_2	<2.10-6	1,1,2,2-C ₂ H ₂ F ₄	<5·10 ⁻⁷
C_2H_4	<2.10-6	CH ₃ Cl	<5·10 ⁻⁷
C_2H_6	<2.10-6	CF ₃ Cl	<5.10-7
C_3H_6	<2.10-6	Si ₂ H ₆ O	<2.10-6
C ₃ H ₈	<1.10-6	$\mathrm{Si_{3}H_{8}O_{2}}$	(1.3±0.4)·10 ⁻⁶
<i>i</i> -C ₄ H ₁₀	<2.10-6	i-Si ₄ O ₃ H ₁₀	<2.10-7
n-C ₄ H ₁₀	<2.10-6	$n-\mathrm{Si}_4\mathrm{O}_3\mathrm{H}_{10}$	<2.10-7
C ₄ H ₈ 1-buten	<2.10-6	Si ₂ OH ₄ F ₂	<7.10-7
C ₄ H ₈ 2-buten	<2.10-6	Si ₂ OH ₅ F	(1.8±0.5)·10 ⁻⁶
$n-C_5H_{12}$	<2.10-6	$Si_3O_2H_6F_2$	<3.10-7
n-C ₆ H ₁₄	<2.10-6	Si ₃ O ₂ H ₇ F	<4·10 ⁻⁷
$n-C_7H_{16}$	<2.10-6	Si ₂ H ₆	<3.10-6
n-C ₈ H ₁₈	<5.10-6	Si ₃ H ₈	<6.10-7
C_6H_6	<1.10-6	i-Si ₄ H ₁₀	<6.10-7
C ₆ H ₅ -CH ₃	<4·10 ⁻⁷	n-Si ₄ H ₁₀	<3.10-7
GeH ₄	<3.10-6	i-Si ₅ H ₁₂	<3.10-6
PH_3	<1.10-6	n-Si ₅ H ₁₂	<3.10-6
AsH ₃	<3.10-6	CH ₃ SiH ₃	<1.10-6
H_2S	<5.10-6	C ₂ H ₅ SiH ₃	<1.10-6
CF ₄	<4·10 ⁻⁷	(SiH ₃) ₂ CH ₂	<3.10-7
CHF ₃	<5.10-6	(CH ₃) ₃ SiH	<5.10-7
C_2F_4	<3.10-7	$(C_2H_5)_2SiH_2$	<6.10-7
1,1,1-C ₂ H ₃ F ₃	<7.10-7	Si ₂ H ₅ CH ₃	<1.10-6

of hydrocarbons, siloxanes, homologues of monosilane, and hydrides of a number of chemical elements) in an isotopically enriched monosilane – rectifier ²⁸SiH₄, according to the chromatography-mass spectrometry data.

Fig. 1 shows an image of a sample of a single-crystal silicon–28 (content of the main isotope 99.9988 wt. %) weighing 90 g.

Table 5 shows the content of germanium isotopes in the composition of samples of isotopically enriched monogermane ⁷²GeH₄ and germanium-72 obtained from it according to inductively coupled plasma mass spectrometry.

From Table 5, it can be concluded that there is no statistically significant isotopic dilution at the stage of obtaining crystalline germanium-72.



Fig. 1. Image of a single crystal silicon sample – 28

Table 6 provides information on the content of a number of impurities (hydrocarbons, halogen derivatives of hydrocarbons, siloxanes, homologues of monosilane, hydrides of a number of chemical elements) in isotopically enriched monogermane – rectifier ⁷²GeH₄ according to chromatography-mass spectrometry data.

Table 5. The content of germanium isotopes in the composition of isotopically enriched monogermane samples 72 GeH, and germanium-72

Cubatanaa	The content of the germanium isotope, wt. %					
Substance	⁷⁰ Ge	⁷² Ge	⁷³ Ge	⁷⁴ Ge	⁷⁶ Ge	
⁷² GeH₄	0.00035±0.00007	99.98460±0.00020	0.01140±0.00015	0.00357±0.00012	0.00008±0.00004	
⁷² Ge	0.00011±0.00002	99.98576±0.00105	0.01099±0.00096	0.00310±0.00030	0.00004±0.00001	

Table 6. The content of molecular impurities in high-purity monohermane ⁷²GeH,

Impurity	Impurity content, mol. %	Impurity	Impurity content, mol. %
Ar	(2.8±0.3)·10 ⁻⁵	$i-C_4H_{10}$	<2.10-6
$_{-}$ CO $_{_{2}}$	(3±1)·10⁻ ⁶	$n-C_5H_{12}$	<1.10-6
N_2O	<2.10-6	$n-C_6H_{14}$	<1.10-6
Kr	<3.10-7	i-C ₇ H ₁₆	<1.10-6
Xe	<3.10-7	C ₇ H ₁₆ 3-metylhexane	<1.10-6
$\mathrm{SiH}_{_{4}}$	<1.10-6	$n-C_7H_{16}$	<9.10-7
PH_3	<1.10-6	$n-C_8H_{18}$	<2.10-6
AsH ₃	<2.10-6	C_6H_6	<2.10-7
H_2S	<3.10-6	$C_6H_5CH_3$	<2.10-7
CH ₄	<1.10-5	C_2H_5C1	<5⋅10⁻⁻
C_2H_2	<2.10-6	CH_2Cl_2	<5⋅10⁻⁻
C_2H_4	<1.10-6	2-C ₃ H ₇ Cl	<9.10-7
C_2H_6	<2.10-6	C_4H_9F	<4.10-6
C_3H_6	<1.10-6	C ₂ H ₅ GeH ₃	<1.10-5
C_3H_8	<1.10-6	$C_2H_5Ge_2H_5$	<1.10-5
C ₄ H ₈ 2-metyl-1-propen	<2.10-6	1,1,2-C ₂ F ₃ Cl ₃	<4.10-7
C ₄ H ₈ 1-buten	<2.10-6	1,1,2-C ₂ H ₃ Cl ₃	<6.10-7
C ₄ H ₈ 2-buten	<2.10-6	CH ₃ GeH ₃	<1.10-6
$n-C_4H_{10}$	<2.10-6	CS_2	<4.10-7

Currently, high-purity isotopically enriched monosilane and monogermane, as well as samples of monoisotopic silicon and germanium with a controlled content of isotopes with a non-zero nuclear spin, produced at the Institute of Chemistry of High Purity Substances of the Russian Academy of Sciences, are widely used in scientific research dedicated to the creation of quantum computing devices in Russia and abroad.

Research on the properties of isotopically enriched binary and complex substances is of fundamental scientific interest. Of particular interest is the production and study of the effect of the isotopic composition on the properties of optical materials, such as quartz glass. A theoretical evaluation provided in a number of works [59, 60] allows concluding that optical losses can be reduced and the optical fiber

transparency window can be expanded by the ³⁰Si¹⁸O₂-based light-conducting core and the ²⁸Si¹⁶O₂-based shell. According to the authors, it is also possible to ensure the effect of total internal reflection in such a quartz light-guiding structure due to the difference in the isotopic composition of silicon and oxygen without the use of alloying additives [4]. Silicon tetrachloride is used to obtain high-purity quartz glass by various techniques (vapor deposition, solgel method), while its tetrafluoride is used to separate silicon isotopes by gas centrifugation. For the synthesis of ²⁸SiCl₄, a technique was developed which involves the conversion of isotopically enriched silicon tetrafluoride in static conditions using aluminum(III) chloride [61]; the isolated yield is about 95%, the productivity is 3-4 g of ²⁸SiCl₄/hour. The reaction of silicon

tetrafluoride with aluminum(III) chloride proceeds successively with the formation of silicon fluoride-chlorides [62]; the equilibrium of liquid-vapor in the system $SiCl_4 - SiCl_{4-n}F_n$ impurities ($n = 1 \div 4$) [63] was also studied. A technique has been developed for the synthesis of silicon-28 tetrachloride from simple substances in a quartz reactor (productivity 20 g/h), which, in combination with isothermal distillation, allows obtaining samples of $^{28}SiCl_4$ with enrichment at 99.99% and the content of impurities of common chemical elements (sodium, aluminum, calcium, iron) at $n \cdot 10^{-5}$ wt. %.

High-purity silicon-28 tetrachloride has been used for the manufacture of preforms and fiber light carriers based on isotopically enriched quartz glass by the MCVD method [64]. Deposition of isotopically enriched silicon dioxide was carried out on a support tube with a natural content of isotopes; isotopically enriched ²⁸SiF, with a degree of enrichment in silicon-28 of 99.99782±0.00012% was used as a substance for doping with fluorine and reducing the refractive index of the light-reflecting shell. This resulted in obtaining a piece with a light-guiding structure based on isotopically enriched quartz glass, from which the light-guide was extracted. The refractive index profile of the obtained piece was measured, as well as the spectral dependence of the level of optical losses of the light guide based on isotopically enriched silicon-28 dioxide in the range of 900–1,750 nm [64]. Figure 2 shows

the distribution profile of silicon isotopes in a preform based on isotopically enriched $^{28}{\rm SiO}_2$ according to secondary ion mass spectrometry.

The content of silicon isotopes 28 Si, 29 Si, and 30 Si in the central part of the preform was 99.89±0.10%, 0.087±0.080%, and 0.023±0.020%, respectively. These results indicate a marked decrease in the content of the isotope of silicon-28 in the preform relative to its content in the starting substances (28 SiCl $_4$ and 28 SiF $_4$) to a level of 99.9%. Most likely this was caused by the diffusion of silicon isotopes 29 Si, 30 Si from a support quartz tube with a natural isotopic composition of silicon at the stage of the porous layer melting.

Further research in the field of obtaining isotopically enriched substances and materials based on them should be focused on increasing the level of isotopic purity and obtaining samples of binary substances enriched in the basic chemical elements in their composition.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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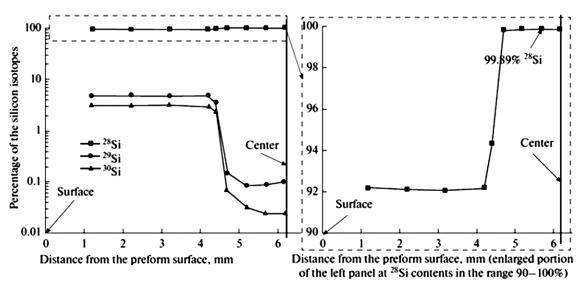


Fig. 2. The distribution profile of silicon isotopes in a preform based on isotopically enriched ²⁸SiO² [64]

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Review

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Advanced methods for preparing especially pure glasses based on germanium and gallium chalcogenides. Part 1. Synthesis via volatile and low-melting compounds. Review

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Abstract

Glasses based on germanium and gallium chalcogenides are promising optical materials for the near and mid-infrared (IR) regions. They are used to develop fiber-optic sensors, sources of supercontinuum, luminescent and laser radiation, glass-ceramic materials with improved mechanical properties, memory cells, and other optical and optoelectronic devices. The most important characteristic of chalcogenide glasses is the content of limiting impurities that have the most negative effect on their optical properties. Conventional methods for producing these materials include melting simple substances with getters in evacuated silica-glass ampoules and then distilling the melt. These methods do not allow achieving extremely low concentrations of impurities that do not affect optical transparency of glasses. Therefore, new approaches need to be developed.

The purpose of the review is to systematize the scientific information related to the methods for preparing especially pure chalcogenide glasses which have been developed over the past 15 years at the Institute of Chemistry of High Purity Substances of the Russian Academy of Sciences. The methods discussed in the first part of the paper include: 1) synthesis of p-element chalcogenides via volatile iodides; 2) preparing a batch by thermal decomposition of germanium sulfide and selenide iodides; 3) synthesis and deep purification of germanium monochalcogenides. The developed methods made it possible to reduce the content of hydrogen, oxygen, and carbon impurities and heterogeneous inclusions in chalcogenide glasses by 1-2 orders of magnitude as compared to conventional methods. In conclusion, the article discusses the possibilities for further reduction of the content of impurities in glasses based on germanium and gallium chalcogenides to achieve extremely low optical losses.

Keywords: Chalcogenide glasses, Especially pure substances, Optical materials, Synthesis, IR spectrometry, Chemical transport reactions

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1. Introduction

Chalcogenide glasses are a promising material for mid-infrared (IR) optics. It is a broad class of glassy inorganic materials based on sulfides, selenides, and tellurides of elements belonging to groups 13-15 of the periodic system. These glasses have a wide transparency region in the IR range (Fig. 1), high optical nonlinearity, and pronounced semiconductor and acoustooptical properties [1–3]. Typical chalcogenide glass forming systems include: As - S, As - Se, As -S - Se, As - Se - Te, Ge - S, Ge - Se, Ge - As - S, Ge - As - Se, Ga - Ge - S, Ga - La - S, Ge - Sb - S, Ga - Ge - Se, Ga - Ge - Te, etc. [4, 5]. There is an increasing interest in glasses based on gallium and antimony chalcogenides, which do not contain arsenic or germanium [6, 7]. This is due to their better transparency in the long-wave IR range.

Chalcogenide glasses are widely used as materials for IR lenses in night vision equipment and thermal visors and for the manufacture of fiber-optic light carriers used in aircraft protection systems [8, 9]. Chalcogenide glasses are used to develop fiber-optic sensors, supercontinuum sources, memory cells, and other optical and optoelectronic devices [10–12]. These materials can also be used to manufacture optical glass-ceramics with improved mechanical and thermophysical properties as compared to glass [13]. The most important area of application

of chalcogenide glasses doped with rare earth elements (REE) is the development of midinfrared luminescent and laser sources [14, 15].

The key characteristic of chalcogenide glasses, which largely determines their practical application, is the content of limiting impurities that have the most negative effect on their optical properties. Among such impurities are: 1) hydrogen in the form of SH-, SeH-, TeH-, and OH-groups and water; 2) oxygen in the form of oxides of glass components and impurity elements; 3) carbon-containing compounds: CS₂, CO₂, COS, and organic substances; 4) transition metals; 5) heterogeneous impurity inclusions, which, first of all, include silicon(IV) oxide used as a material for glass synthesis equipment. These impurities have intense absorption bands within the region of optical transparency of chalcogenide glasses (Table 1) [16-19]. Heterogeneous inclusions reduce the overall transparency in a wide spectral range due to the scattering of IR radiation [20]. The influence of heterogeneous inclusions on the optical transparency of glasses is mainly determined by their size. For $Ge_{25}Sb_{10}S_{65}$ glass, it was shown that with a SiO, particle size of 0.1 µm and a concentration of 10⁹ cm⁻³ optical losses in the fiber are at least 1 dB/m in the spectral range of 1.1-7.4 µm. Particles with a size of 1 µm cause the same losses but at concentrations of 10⁵ cm⁻³ [21].

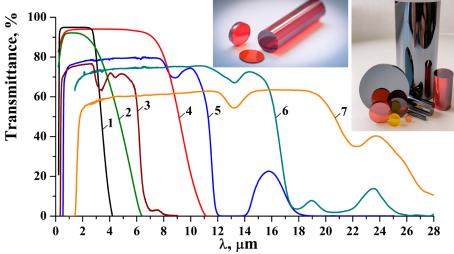


Fig. 1. Transparency windows of several optical materials in the near and middle IR ranges. 1 – silica glass; 2 – Al_2O_3 ; 3 – ZnO– TeO_2 tellurite glass; 4 – CaF_2 ; 5 – $Ga_5Ge_{35}S_{60}$ sulfide glass; 6 – $Ge_{28}Sb_{12}Se_{60}$ selenide glass; 7 – $(GeTe_4)_{50}(AgI)_{50}$ telluride glass. An optical path-length of samples can be different, therefore a comparison of transparency spectra is estimate. Inserts give photos of chalcogenide glasses prepared at the Institute of Chemistry of High-Purity Substances of the Russian Academy of Sciences

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Table 1. The position of the absorption bands of impurities, λ_{\max} ; their absorption coefficients, ϵ ; and their limiting contents in $\mathrm{As_2S_3}$ glass corresponding to the intrinsic optical losses, C_{\lim} [16–19, 24]

Impurity	λ _{max} , μm	ε, dB/km/ppm(wt.)	C _{lim} , ppb(wt.)	Impurity	λ _{max} , μm	ε, dB/km/ppm(wt.)	C _{lim} , ppb(wt.)
SH	4.0	2,500 ppm(at.)	0.4	SO ₂	8.63	43	2300
SeH in As ₂ Se ₃ glass	4.5	1,000 ppm(at.)	0.1	Se-O in As ₂ Se ₃ glass	10.6	380	2.6
GeH	4.85	_	_	Te-O	13.1	_	_
OH (in SiO ₂)	2.92	104	0.3	Ga – O	15-20	_	_
Ge-O in	7.8	2610	0.4	Si-O	9.1	28000	_
$Ge_{25}Se_{15}Te_{60}$ glass	12.5	99000	0.01	CO ₂	4.33	1.5·10 ⁴	0.04
As_2O_3 in As_2Se_3	12.65	4.3·10 ⁴	100	COS	4.95	10 ⁵	0.004
glass	9.5	1030	400	CS ₂	6.68	4.8·10 ⁵	0.2

The intrinsic (excluding the effect of impurities) optical losses in As₂S₃ glass, calculated from the absorption spectra of an ultra-pure bulk sample, are <20 dB/km in the spectral range of 2–6 μm with a minimum of about 0.3 dB/km at 5.2 µm [22]. According to the theoretical evaluation, the minimum optical losses for As, Se, glass in the range of 2.5–6.6 µm do not exceed 1 dB/km [23]. From the known absorption coefficients of impurities [24], it was calculated that their content in As₂S₃ glass, corresponding to the intrinsic optical losses, should not exceed 0.4 ppb(wt.) for hydrogen in the form of SH-groups and 0.2, 0.004, and 0.04 ppb(wt.) for carbon in the form of CS₂, COS, and CO₂, respectively. To provide minimum optical losses in As, Se, glass, the content of hydrogen impurity in the form of SeH-groups should not exceed 0.1 ppb(wt.), of oxygen in the form of oxides 0.1 ppm(wt.), and carbon 0.1–0.01 ppm(wt.) [25]. The content of impurities in samples of chalcogenide glasses prepared by the conventional method including the purification of the melt by distillation is 3-4 orders of magnitude higher than the permissible values [16]. This significantly limits the practical use of such materials for the manufacture of fiberoptic light carriers and makes it impossible to use them to manufacture laser radiation sources. The intrinsic optical losses in glasses based on germanium and gallium chalcogenides have so far been only evaluated for the Ge_zPS_{7,5} composition [26], which makes it difficult to determine the maximum permissible concentration of impurities in these materials.

The purpose of this review is to systematize the scientific information related to the methods for

preparing ultra-pure chalcogenide glasses which have been developed over the past 15 years at the Institute of Chemistry of High Purity Substances of the Russian Academy of Sciences. The key idea of the new methods was to use germanium and gallium compounds, which are more volatile, low-melting, and have better reactive properties as compared to the simple substances which had been used traditionally. Among such compounds were: germanium(IV) iodide, gallium(III) iodide, germanium sulfide and selenide iodides, and germanium(II) chalcogenides.

2. Traditional method for preparing chalcogenide glasses

Traditionally, chalcogenide glasses are prepared by the interaction of especially pure substances of grades 4–7 N in evacuated silicaglass ampoules [27]. The synthesis temperature is determined by the glass composition and is usually in the range of 750–950 °C. To ensure the uniformity of the glasses, the process is carried out in rocking furnaces. The melt is usually quenched in air or in water, depending on the crystallization stability of the glasses. To relieve mechanical stresses after quenching, the samples are annealed at the glass transition temperature.

The main method of reducing the content of impurities in chalcogenide glasses involves adding getters to the batch, which selectively bind impurities and convert them into a form which can be easily removed, and a subsequent purification of the melt by distillation [27]. Aluminum and magnesium is used to bind oxygen impurities, while tellurium(IV) chloride and aluminum(III) chloride

contribute to a decrease in hydrogen content. This method allows reducing the concentration of impurities in glasses by 1-2 orders of magnitude. In the best samples of germanium-containing glasses prepared by the traditional method, the content of hydrogen impurity in the form of SH-groups was 7.5 ppm(at.) in $\text{Ge}_{25}\text{Sb}_{10}\text{S}_{65}$ [17]; the content of hydrogen impurity in the form of SeH-groups was 2.7 ppm(at.) and the content of oxygen in the form of germanium oxides was 0.2 ppm(wt.) in $\text{Ge}_{20}\text{Se}_{80}$ [28].

The disadvantage of using getters is their interaction with the walls of the silica-glass reactor, which leads to the introduction of impurities into the chalcogenide melt.

$$2Al + 3SiO_2 \rightleftharpoons Al_2O_3 + 3SiO, \tag{1}$$

$$xAl_2O_3 + ySiO_2 \rightleftharpoons xAl_2O_3 \cdot ySiO_2,$$
 (2)

$$4Mg + SiO_2 \rightleftharpoons 2MgO + Mg_2Si.$$
 (3)

During subsequent purification of the glass-forming melt by distillation, reaction products (1)–(3), which have low solubility in the chalcogenide melt, can be only partly removed. This results in the appearance of heterogeneous impurity inclusions and absorption bands in the glasses in the regions of 9.1 μ m (Si–O) and 13.2 μ m (Al-O) [19]. The interaction of aluminum with the walls of the silica-glass reactor can lead to its rupture during cooling of the melt. This significantly complicates the production of chalcogenide glasses and makes synthesis more hazardous.

The main disadvantage of the purification of the batch by distillation is the high temperature and duration of the process. They are due to the relatively low volatility of germanium dichalcogenides and the need for slow evaporation of the melt for its effective purification from heterogeneous impurities. High temperatures contribute to the entry of hydrogen, metal, and silicon oxide impurities from the walls of the silica-glass reactor into the chalcogenide melt [22, 29]. During preparation of telluride-germanium glasses, there is another limitation of purification by distillation, i.e. the dissociative evaporation of GeTe, which makes it difficult to obtain charge of a specified chemical composition [30].

The main challenge for the production of especially-pure glasses containing gallium is related with the low volatility of this metal and its chalcogenides [31, 32]. This makes

the purification of the glass-forming melt by distillation impossible. Therefore, gallium is added to the batch by loading the sample in air or into a glove box with an inert atmosphere. When using this method, gallium oxide and hydroxide impurities, which are present on its surface and which form during the reactor soldering, enter the glass-forming melt. These impurities significantly reduce the transparency of chalcogenide glasses in the mid IR range.

These problems associated with the production of optical materials based on especially pure chalcogenide glasses mean that novel approaches to their synthesis need to be developed. Reducing the content of absorbing and scattering impurities by 2–3 orders of magnitude will significantly improve the optical characteristics of chalcogenide glasses. This will expand the scope and the efficiency of their application. To improve the purity of chalcogenide glasses, it is highly important to establish the nature and limits of the influence of impurities on the properties of these materials.

3. Preparation of chalcogenide glasses via volatile iodides

The main idea of the developed method was to synthesize batch components by the interaction between iodides and chalcogen [33–35]. Using the Ge-Sb-S-I and Ge-Sb-Se-I systems, germanium and antimony chalcogenides were synthesized by reactions of the following type:

$$GeI_4 + 2S(Se) \rightleftharpoons GeS(Se)_2 + 2I_2\uparrow,$$
 (4)

$$2SbI_z + 3S(Se) \rightleftharpoons Sb_2S(Se)_z + 3I_2\uparrow. \tag{5}$$

According to the thermodynamic modeling of reactions (4) and (5), the degree of iodides conversion to the corresponding chalcogenides under equilibrium conditions was expected to be low, up to 26% for sulfides and up to 11% for selenides at a temperature of 500 °C [36]. A significant increase in the yield of chalcogenides was possible due to the selective isolation of iodine from the reaction melt. In practice, this involved using a vertical unit made of silica-glass and consisting of three parts which were successively soldered together: 1) reactor; 2) mass transfer section; 3) iodine receiver. Batch components were synthesized at a temperature of 500–650 °C. Heating of the reaction mixture resulted in the

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release of iodine vapors by reactions (4) and (5) and their entering the receiver. One part of the iodine condensed as crystals, the other part in the form of liquid drained back into the reactor. The presence of the mass transfer section allowed selectively extracting iodine from the reaction melt without the significant removal of volatile iodides from it. As the synthesis proceeded, the batch was depleted of iodine and its viscosity increased, which required a gradual increase in the reactor temperature to ensure the batch was molten.

The content of metal impurities in the prepared glass samples did not exceed 0.5 ppm(wt.); the content of silicon, which was the main source of heterogeneous inclusions, was 0.02–0.1 ppm(wt.); the content of the impurities which were the most difficult to remove, impurities of hydrogen in the form of SH- and SeH-groups, was 0.1–0.3 ppm(at.) [33, 34]. This was 1–2 orders of magnitude lower than in glasses prepared by the interaction of simple substances with purification of the melt by distillation. The achieved purity was provided by the following main advantages of the method:

– Before the synthesis of the batch component, iodides of *p*-elements were subjected to deep purification from hydrogen, oxygen, and carbon impurities and heterogeneous inclusions by vacuum distillation at temperatures that did not exceed 200 °C. The content of these impurities in the corresponding simple substances of especially pure grades was not controlled and could be significant.

- Iodides were loaded into the reactor by vacuum evaporation, which excluded the contact of reagents with atmospheric impurities (oxygen, water, organic substances, dust, etc.).
- The synthesis temperature of the glassforming components of the batch did not exceed 650 °C, which significantly reduced the contamination effect of the silica-glass reactor.
- Iodides of *p*-elements converted the hydrogen impurity in the form of SH- and SeH-groups into an easily removable form due to the following chemical reactions:

$$GeI_4 + 2H_2S(Se) \rightleftharpoons GeS(Se)_2 + 4HI\uparrow,$$
 (6)

$$2SbI_3 + 3H_2S(Se) \rightleftharpoons Sb_2S(Se)_3 + 6HI\uparrow,$$
 (7)

$$I_2 + H_2S(Se) \rightleftharpoons S(Se) + 2HI\uparrow.$$
 (8)

The formed hydrogen iodide was removed from the reaction melt and concentrated in the iodine receiver.

The main disadvantage of the method is the difficulty of providing a specified chemical composition of the glass by a controlled removal of iodine from the reaction mixture. Deviations in the content of components can reach 3–5 at. %.

The developed method was used for the synthesis of crystalline binary and complex sulfides and selenides of p-elements with the actual yield of 70–98% and a residual iodine content of <0.1 to 2 at. %. [37–40] (Table 2). Such amount of iodine did not have a noticeable negative effect on the optical properties of chalcogenides, but allowed growing their single crystals by chemical transport directly in the synthesis reactor [42]. This

Table 2. Chemical composition and actual yield of some chalcogenides prepared via volatile iodides

Sample		Yield, %			
Sample	<i>p</i> -element	Zn	S(Se)	I	1 ieiu, 76
Ga ₂ S ₃	40.2	-	59.7	0.04	95
In_2S_3	39.1	-	59.4	1.5	72
Sb_2S_3	38.2	-	61.2	0.6	88
GeS_2	33.9	-	65.9	0.2	92
ZnS	_	50.5	48.7	0.8	91
Ga ₂ Se ₃	41.3	_	57.3	1.4	82
In ₂ Se ₃	42.3	_	56.1	1.6	70
GeSe ₂	36.4	_	63.4	0.2	90
ZnSe	_	50.9	48.3	0.8	90
ZnGa ₂ S ₄	29.2	14.8	55.6	0.4	94
ZnGa ₂ Se ₄	26.9	16.2	56.5	0.4	88

means there was no need in the extra step which involves adding a transport agent and leads to the contamination of the material. It was theoretically predicted and experimentally confirmed that the yield of chalcogenides of p-elements increases with a decrease in the atomic mass of the element in the group and with the transition from selenides to sulfides. Synthesis was most effective when aluminum(III) iodide and gallium(III) iodide interacted with sulfur. To prepare Ga₂S₃ and complex sulfides based on it, a reactor with two temperature zones without a mass transfer section was used. At synthesis temperatures of up to 550 °C, γ-Ga₂S₃ with the cubic structure of the unit cell was formed (Fig. 2a). Annealing the resulting sulfide at higher temperatures led to the formation of monoclinic β -Ga₂S₂. The synthesis of indium, germanium, antimony, and bismuth chalcogenides was complicated by the formation of relatively stable sulfide and selenide iodides as intermediate products (Fig. 2b, 2c). The decomposition of these compounds led to the formation of chalcogenides, for example, by reactions:

$$3SbSI \rightleftharpoons Sb_2S_3 + SbI_2, \tag{9}$$

$$Bi(Bi_{2}S_{2})_{0}I_{2} \rightleftharpoons 9Bi_{2}S_{2} + BiI_{2}.$$
 (10)

Preparing tellurides of p-elements of groups 13–15 by the proposed method is complicated due to thermodynamic limitations with regard to the high melting point of tellurium ($T_{\rm m}$ = 451 °C [43]) and the formation of relatively stable TeI₄.

4. Preparation of glasses of the Ge – S(Se) – I system by thermal decomposition of germanium sulfide and selenide iodides

It is known that the Ge – S – I system has low-melting compounds of the GeSI $_2$ (T_m = 150 °C) [44] and Ge $_2$ S $_3$ I $_2$ (T_m = 310 °C) composition [45], which form stable glasses. Thermal decomposition of these sulfide iodides in a temperature gradient was accompanied by the release of germanium(IV) iodide and the formation of germanium(IV) sulfide [46] (Fig. 3), which in a simplified way can be described by the following chemical reactions:

$$2\text{GeSI}_2 \rightleftharpoons \text{GeS}_2 + \text{GeI}_4 \uparrow,$$
 (11)

$$2Ge_2S_3I_2 \rightleftharpoons 3GeS_2 + GeI_4\uparrow.$$
 (12)

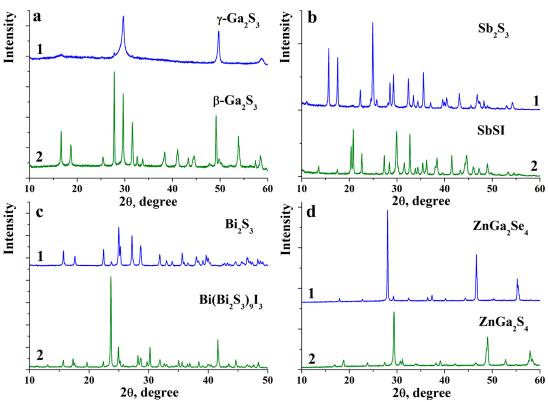


Fig. 2. X-ray diffraction patterns of the reaction products of some iodides with chalcogens; a1 – GaI_3 + S <550 °C (PDF 49-1361 [41]); a2 – GaI_3 + S >550 °C (PDF 76-0752); b1 – SbI_3 + S (PDF 75-1310); b2 – SbI_3 + S intermediate product (PDF 74-2244); c1 – BiI_3 + S (PDF 84-0279); c2 – BiI_3 + S intermediate product (PDF 73-1157); d1 – ZnI_2 + GaI_3 + Se (PDF 47-1590); d2 – ZnI_2 + GaI_3 + S (PDF 84-2007)

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By controlling the degree of decomposition of sulfide iodides, it is possible to specify the composition of the non-volatile reaction product. The method for preparing glasses of the Ge – S – I system at low temperatures is based on this idea [47, 48].

The synthesis of the glassy $Ge_2S_3I_2$, its subsequent thermal decomposition, and melting of the resulting batch were carried out in one evacuated silica-glass reactor, which excluded contamination of the melt with impurities from the atmosphere [47]. $Ge_2S_3I_2$ was prepared by the reaction between germanium(IV) iodide, germanium, and sulfur at a temperature of 550 °C

$$GeI_4 + 3Ge + 6S \rightleftharpoons 2Ge_2S_2I_2.$$
 (13)

When a temperature gradient from 550 °C to room temperature was created in the reactor, the resulting sulfide-iodide was thermally decomposed. The degree of decomposition was controlled by its duration (from 1 to 5 hours) according to a predetermined experimental dependence of the glass composition on the duration of the process. This allowed specifying the content of components with deviations of no more than ± 2 at. %. Further,

the evaporator with the resulting batch was sealed-off from the unit and the glass-forming melt was homogenized at 750 °C under the stirring action of a rocking furnace. Similarly, glasses of the Ge – Se – I system were prepared by the thermal decomposition of Ge₂Se₃I₂ [49].

According to the results of the IR Fourier analysis, in the best samples of the prepared glasses, the content of hydrogen impurity in the form of SH- and SeH-groups was 0.1-0.5 ppm(at.); OH-groups <0.05 ppm(wt.); oxygen chemically bound to germanium up to 0.5 ppm(wt.), and carbon in the form of CO_2 , CS_2 , and COS < 0.01 ppm(wt.). The achieved content of absorbing impurities was due to a significant decrease in the synthesis temperature and effective removal of highly volatile impurity compounds at the stage of thermal decomposition of $Ge_2S_3I_2$ and $Ge_2Se_3I_2$.

The main limitation of the developed method is that it can only be applied to two glass-forming systems: Ge - S - I and Ge - Se - I. To prepare glasses of the compositions beyond the $GeS(Se)_2 - GeI_4$ quasi-binary section, it is necessary to use source alloys with a corresponding excess (y) or deficiency (-y) of chalcogen:

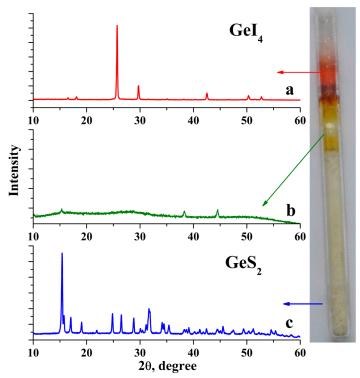


Fig. 3. X-ray diffraction patterns of the thermal decomposition products of $Ge_2S_3I_2$ glass. a – cold section of the reactor, corresponds to GeI_4 (PDF 75-0982); b – intermediate section, amorphous product with weak reflections from GeS_2 ; c – high-temperature section, corresponds to GeS_2 (PDF 71-0003). The insert on the right shows a photo of the reactor

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$$2Ge_2S_{(3+v)}I_2 \rightleftharpoons 3GeS_{(2+2/3v)} + GeI_4\uparrow.$$
 (14)

Along with the use of sulfide and selenide iodides, the glasses of the Ge – S and Ge – Se systems can be prepared by thermal decomposition of $GeSBr_2$ and $Ge_2S_3Br_2$ [50]. The disadvantage of this method is the need to use liquid bromine to synthesize starting compounds.

5. Preparation of glasses via germanium monochalcogenides

5.1. Ge - S, Ge - Se systems

Another approach was aimed at alleviating the disadvantages of the conventional vacuum distillation of the chalcogenide melt described in paragraph 2. The novel method involved synthesis and purification of germanium monochalcogenides. The synthesis was carried out by passing chalcogen vapors over a mixture of germanium granules with a getter [51, 52]. The by-products of the proposed method for the germanium monochalcogenide synthesis were dichalcogenides. According to the results of thermodynamic modeling of the Ge - S and Ge - Se systems by the method of equilibrium constants, the formation of GeS and GeSe is facilitated by an increase in the reactor temperature and a decrease in the temperature of the chalcogen evaporator [51, 53]. The results of thermodynamic modeling are in good agreement with the experimental values of the optimal conditions for the synthesis of germanium monochalcogenides: reactor temperature of 550-600 °C and chalcogen evaporator temperature of 240–260 °C for sulfur and 330–350 °C for selenium.

The developed method was used to prepare especially pure glasses of the Ge₂₀Se₈₀ composition [54]. The content of impurities in the optical fibers fabricated from such glass was as following: hydrogen in the form of SeH-groups was 1.8±0.1 ppm(at.); oxygen chemically bound to germanium was 0.013±0.002 ppm(wt.). According to these values, the obtained fibers were the best among those made of glasses based on germanium selenides. The minimum optical losses in the fiber were 0.77 dB/m at a wavelength of 5.6 µm (Fig. 4). The relatively high losses may be due to partial crystallization of the glass due to unoptimized conditions for fiber drawing. The lowest losses in germanium selenide optical fibers - obtained to date [28] are 0.1 dB/m at a wavelength of 6.5 µm. The content of hydrogen impurities in the form of SeH-groups and oxygen chemically bound to germanium in this fiber was 2.7 ppm(at.) and 0.2 ppm(wt.), respectively.

For sulfide systems, the developed method in combination with the synthesis of gallium(III) sulfide by the interaction of gallium(III) iodide with sulfur was used to especially pure Ga₅Ge₃₅S₆₀ and Ga₈Sb₃₂S₆₀ glasses [53]. The absorption spectra of the samples and the photographs of diffraction spots caused by submicron inclusions recorded by 3D laser ultramicroscopy are shown in Fig. 5. The content of hydrogen impurities in the samples in the form of SH-groups was 0.2-0.6 ppm(at.), which was 2 orders of magnitude lower than during the synthesis from simple substances. In the glasses prepared by the conventional method, the content of heterogeneous micron-sized inclusions determined by optical microscopy was $n \cdot 10^3$ pcs/cm³. The content of submicron inclusions (0.1-0.9 µm) recorded by 3D laser ultramicroscopy reached $n \cdot 10^8$ pcs/cm³. The samples prepared by the developed method had single micron inclusions (<10² pcs./cm³). In the submicron region, there was only background scattering, i.e. intrinsic scattering on frozen fluctuations or on small (<100 nm) inclusions that did not contribute significantly to optical losses in the mid-infrared range.

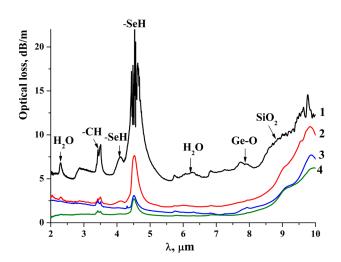


Fig. 4. Spectra of total optical losses of optical fibers without a reflective cladding, made of $Ge_{20}Se_{80}$ glasses prepared: from simple substances without additional purification (1); with the addition of aluminum to the batch and double distillation of the melt (2); by passing selenium vapor over germanium (3); by passing selenium vapor over a mixture of germanium and aluminum (4) [54]

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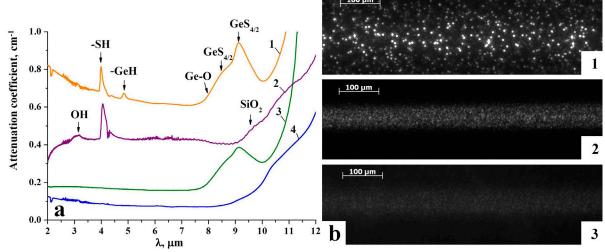


Fig. 5. Absorption spectra (a) and photographs of diffraction spots from submicron inclusions (b) of $Ga_5Ge_{35}S_{60}$ and $Ga_8Sb_{32}S_{60}$ glasses; $Ga_5Ge_{35}S_{60}$ samples prepared by the traditional method (a1, b1) and by the developed method (a3, b2); $Ga_8Sb_{32}S_{60}$ samples prepared by the traditional method (a2) and by the developed method (a4, b3) [53]

The developed method has the following key advantages as compared to the conventional distillation of the melt:

- Higher volatility of germanium monochalcogenides as compared to dichalcogenides allows reducing the temperature of the vacuum purification of the batch by 50–100 °C.
- The addition of the getter to the intermediate ampoule with germanium minimizes the products of its interaction with impurities and the silicaglass reactor entering the glass-forming melt.
- Passing chalcogen vapors over the mixture allows for preliminary chemical etching of germanium granules before the batch components are loaded into the reactor. This contributes to the removal of surface contamination from the granules (adsorbed gases, water, OH-groups, oxides, and heterogeneous inclusions) due to the capture of germanium monochalcogenide by the vapors or due to chemical reactions of type (15)–(17):

$$3S + GeO_2 \rightleftharpoons GeS_2 + SO_2,$$
 (15)

$$3S + 2H_2O \rightleftharpoons 2H_2S + SO_2, \tag{16}$$

$$(y/2+2x)S + C_yH_y \rightleftharpoons y/2H_2S + xCS_2. \tag{17}$$

For this, the first portions of germanium monochalcogenide ($50-100\,\mathrm{mg}$), in which surface impurities can be concentrated, are removed from the system.

5.2. The Ge - Te system

Another limitation of the conventional vacuum distillation of the batch during the

preparation of glasses based on the Ge-Te system (for example, $Ge_{20}Te_{80}$, $Ga_{10}Ge_{15}Te_{75}$) is the dissociative evaporation of germanium monotelluride [30].

$$(x+y+z)$$
GeTe_(кр.) \rightleftharpoons $\underset{\sim}{\rightleftharpoons} x$ GeTe_(г.) + $y/2$ Te₂ + $z/2$ GeTe_{2(г.)} + $(y+z/2)$ Ge_(кр.). (18)

The released germanium remains in the evaporator, which leads to a noticeable deviation of the glass composition from the specified value. According to the experimental results, after single, double and triple distillation of the melt, the deviations in the content of germanium in Ge₂₀Te₈₀ glasses reached 0.5, 0.9, and 2.1 at. % [55]. This corresponds to the degree of GeTe decomposition of 2.5, 4.5, and 10.5 %, respectively. In the developed method, the problem of preserving the chemical composition of the batch is solved due to the constant presence of tellurium vapors over GeTe, which shifts the equilibrium of reaction (18) to the left. However, re-distillation of the $(GeTe)_x Te_{100-x}$ batch aimed at further reduction in the content of impurities was accompanied by a noticeable decomposition of GeTe. This could be explained by the fact that at the initial stage of low-temperature distillation tellurium evaporates completely. Suppression of the germanium(II) telluride dissociation caused by excess chalcogen in the vapor phase is impossible during the final stages of purification.

To solve this problem, a novel method for purification of the $(GeTe)_x Te_{100-x}$ batch by distillation was developed [55]. The main idea of

the proposed method is to provide for the separate condensation of tellurium and germanium(II) telluride during the purification of the glassforming melt by distillation. The separation of the batch components allowed for its subsequent purification without GeTe dissociation. This was achieved by the possibility of sequential distillation of GeTe and tellurium from separate ampoules. During the final stages of the germanium(II) telluride evaporation, tellurium vapors were passed over it, which suppressed reaction (18) in the forward direction. In the case of partial decomposition of GeTe during distillation, tellurium vapors interacted with the released germanium and transferred it to the vapor phase. This eliminated the incomplete transition of germanium from the evaporator to the receiver during the purification of the (GeTe), Te_{100-x} batch.

The deviations of the compositions of $Ge_x Te_{100-x}$ glasses prepared by the developed method of purification did not exceed 0.2 at. % even after triple distillation of the melt. The corresponding degree of the GeTe decomposition was 1.0%. Such deviations do not have any noticeable effects on the key properties of chalcogenide glassy materials. This method is an integral part of comprehensive approaches to the production of especially pure glasses based on the Ge-Te system.

6. Conclusions

The developed methods allow significantly (by 1–2 orders of magnitude) reducing the content of impurities in glasses based on germanium and gallium chalcogenides. This is largely due to the fact that the methods implement the chemical principles of the deep purification of substances. As a rule, the efficiency of a single stage of purification from impurities using chemical reactions is higher than that using physicochemical methods (distillation, crystallization, etc.) [56]. In chemical methods, the partition coefficient is determined by the difference in the thermodynamic potentials of reactions with the participation of the main substance and impurity. This difference can significantly exceed the differences in the thermodynamic functions of the phase transitions of the impurity and the purified substance (melting, evaporation), on which the physicochemical methods of separation are based. In turn, these differences are determined by the fact that in chemical methods the purification effect is due to the re-distribution of intramolecular bonds, the energy of which, in general, is higher than that of intermolecular bonds. The efficiency of separation in chemical methods depends on the type of reaction. By selecting the appropriate reagent, it is possible to provide high values of the partition coefficients. A definitive advantage of chemical methods is their applicability in cases when the main substance is a non-volatile, high-melting or thermally unstable compound. Nonetheless, to prepare chalcogenide glasses with a low content of all types of impurities, it is necessary to use a set of methods with regard to the purification and loading of the batch components into the reactor, including vacuum distillation.

An important technique implemented in the developed methods is the transformation of the chemical form of the batch components in order to increase their volatility and meltability and to change (increase or decrease) their reactivity properties. This allows: 1) significantly reducing the temperature and duration of the synthesis; 2) reducing the interaction with the silicaglass reactor; 3) increasing the efficiency of the removal of impurities.

The main disadvantage of chemical methods of purification is the contamination of the main substance with the elements present in the reagents. In the developed methods, the key reagents are iodides. At moderate concentrations iodine does not have any noticeable negative effects on the target properties of chalcogenide glasses. Moreover, it is known that the addition of iodine increases the crystallization stability of glasses, increases their overall transparency and expands its range, and contributes to the dissolution of REE [57–60]. This makes iodides of *p*-elements one of the most suitable transport agents for the production of especially pure chalcogenide glasses.

Despite a significant increase in purity, the achieved content of impurities is still 2 orders of magnitude higher than the upper values, which do not affect the optical properties of chalcogenide glasses. This means that it is necessary to further improve the developed methods, their temperature and time regimes, and equipment. To reduce the contamination effect of the silica-glass reactor

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during chemical transport with the participation of iodides, individual elements or coatings can be developed for the zones of the highest temperature. These elements or coatings can be made of materials such as corundum, pyrolytic boron nitride, etc. The original silica-glass tubes should meet high quality standards in terms of the content of impurities of OH-groups and metals. Surface defects should be under strict control since they are the most chemically active centers when the reactor walls come into contact with the chalcogenide melt. It is necessary to develop methods for reducing the content of these impurities and defects in reactors for thermal, chemical, plasma-chemical, and other types of treatment [61,62]. An important area for further research is the assessment of the minimum achievable optical losses in glasses based on germanium and gallium chalcogenides to determine the limits of the influence of impurities on this key property.

Author contributions

A. P. Velmuzhov: research concept, methodology development, conducting research, text writing, final conclusions. M. V. Sukhanov: research concept, methodology development, conducting research, editing. E. A. Tyurina: methodology development, conducting research, editing. V. S. Shiryaev: research concept, conducting research, editing.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Phases with layered (AB) and "defective" (A_2B_3) structures in $A^{III}-B^{VI}$ systems. Part 2. Phase diagrams and approaches to some problems of reproducible synthesis in $A^{III}-B^{VI}$ systems. Review

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Abstract

The paper analyzes phase diagrams of $A^{III}-B^{VI}$ systems and phase equilibria involving crystalline compounds formed in these systems. The location of each solid binary phase mainly related to mono- and sesquisulfides families and selenides of aluminum, gallium, and indium on T-x-diagrams is discussed in detail. The homogeneity regions of these phases were also analyzed if the necessary data were available. For polymorphic (or close to them) transformations, the nature of the occurring structural transformations was described and the temperature stability of various modifications of similar composition was analyzed. Using examples of several systems, it was shown how, by changing the experimental conditions, it is possible to reproducibly obtain compounds with the required structure (even for different polytypes of structures with very similar structures of individual layers) and the required composition (including those within the regions of phase homogeneity). Various methods of reproducible inorganic synthesis were considered, taking into account the features of the phase diagram and phase equilibria. In conclusion, current and partially still unresolved issues concerning the characteristics of the $A_2^{III}B_2^{VI}$ compounds were analyzed.

Keywords: Chalcogenides, A(III)B(VI), Stoichiometric vacancies, Phase diagrams, Phase equilibria

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Phases with layered (AB) and "defective" (A_2B_2) structures in $A^{III} - B^{VI}$ systems. Part 2...

1. Introduction

This series of review articles is devoted to the description of two large groups of inorganic substances with very unusual structures and properties: mono- and sesqui-chalcogenides of Al, Ga, and In compounds. The first part of the study considers the structural features of phases of $A_1^{III}B_1^{VI}$ and $A_2^{III}B_3^{VI}$ families and the possible areas of practical use of these compounds based on structural features. This second part of the study aimed to summarize and reconcile the data on phase diagrams and equilibria in A^{III}-B^{VI} systems. The main attention was paid to the relationship between a specific phase diagram and the conditions of formation of certain modifications of crystalline phases of the families of mono- and sesquisulfides and selenides of aluminum, gallium, and indium. In this part of the study we also considered the basics of new methods for precision variation of both the phase and non-stoichiometric composition of already synthesized crystalline chalcogenides. The search for such methods is due to the fact that traditional methods of regulating the composition of the solid phase through vapor (for example, during directional crystallization with controlled vapor pressure) are almost unsuitable for A^{III}B^{VI} chalcogenides due to the very low values of saturated vapor pressures above condensed phases even under conditions of contact with melts (S-L-V) equilibrium) [1]. Both the previous and the present (second) parts of the review are based on the analysis of both third-party literary data and the results of the studies of authors of this review and other VSU employees.

2. Al-S and Al-Se systems

2.1. The T-x-diagram and phase equilibria in the Al-S system

When mentioning the aluminum-sulfur system, any chemist will immediately recall the

existence of only one compound, Al₂S₃, which is easily obtained by direct synthesis (for example, by igniting a mixture of powdered Al + S). Until recently, this aluminum sulfide was considered unpromising for high-tech materials science. Indeed, Al₂S₃ is easily destroyed in air and liquid water (the latter interaction is a well-known convenient method for obtaining gaseous hydrogen sulfide). Despite the above, the example of the Al - S system clearly demonstrates how the phase diagram can become the basis for the creation of a new and possibly very promising chemical technology even in the case when solid intermediate phases cannot be obtained at room temperature (as AlS), or they quickly degrade in the atmosphere (as $Al_{2}S_{3}$).

The detailed analysis clearly demonstrates that the Al – S system is not so simple. Thus, back in 1940, Murakami and Shibata [2] showed that, along with ${\rm Al}_2{\rm S}_3$, there is also **mono**aluminum sulfide AlS, which can only be obtained at temperatures around 1000 °C. However, the region of existence of AlS (~50 °C) can be easily "skipped", since when the upper limit of existence is exceeded, aluminum monosulfide decomposes according to a peritectic reaction (see Fig. 1):

$$AlS = L_2 + \gamma - Al_2 S_3 \tag{1}$$

As the temperature decreases, the AlS phase also decomposes. Moreover, it has not yet been possible to preserve aluminum monosulfide even with rapid quenching. The decomposition of AlS during cooling results in almost pure molten aluminum (L_1 liquid) and aluminum sesquisulfide [3]:

$$AlS = L_1 + \gamma - Al_2 S_3. \tag{2}$$

The authors of later studies [4-8] conducted a number of experiments using thermal and X-ray diffraction analysis (XRD), as well as optical metallography. The results of these studies demonstrated that aluminum monosulfide AlS melts peritectically at ~1058 °C (reaction 1) and ${\rm Al}_2{\rm S}_3$ sesquisulfide melts congruent at ~ 1073 °C. The general appearance of the diagram is shown in Fig. 1, and the most important data on phase equilibria are summarized in Table 1.

Among the features of the presented diagram, in addition to the already mentioned AlS, observed on *T-x*-diagram in a narrow temperature

^{*} Various $A^{III}B^{VI}$ modifications, related by solid-phase transformations, are often characterized by similar but different compositions (for example, $\gamma Ga_{2+\delta}S_3$ and $\alpha' Ga_2S_3$). This contradicts the classical definition of polymorphism, according to which different structures must have the same composition. Taking into account the proximity of structural $A_2^{III}B_3^{VI}$ motifs we discussed phases of the sesqui-chalcogenide families (Ga_2S_3 , Ga_2Se_3 etc.), trying to avoid using the term "polymorphism". We also discussed phases of the monochalcogenide families.

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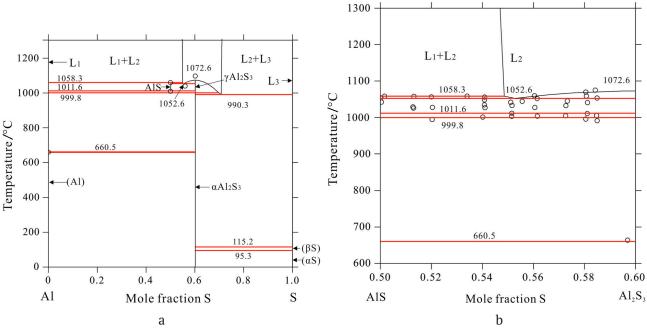


Fig. 1. *T-x*-diagram for the Al – S system [3] (a) and an enlarged fragment of this diagram in the region for a sulfur content from 50 to 60 mol. % (b) [3] (b). The results of the CALPHAD calculation using the substitution solution model are given together with the experimental data

range, two modifications of $\mathrm{Al_2S_3}$ should be noted (α and γ ; the latter melts congruently), as well as three liquid phases separated by wide areas of stratification: L_1 - almost pure molten aluminum, L_3 – almost pure liquid sulfur and L_2 – a melt close in composition to the $\mathrm{Al_2S_3}$. In addition, the authors of a new detailed analysis of the Al – S system [3] came to the conclusion that the upper limit of the existence of aluminum monosulfide is associated not with the peritectic (1), but with a very rare among phase equilibria synthetic transformation, in which a solid phase can be obtained by cooling two immiscible melts (shift to the right of equilibrium 3):

$$AlS = L_1 + L_2 \tag{3}$$

Structurally, AlS monosulfide has not yet been studied. The reason for this is the high aggressiveness of "low-valent" aluminum compounds in relation to almost any solid materials and the impossibility of hardening AlS without decomposition. Several polymorphic modifications have been identified for Al_2S_3 , but, according to the data of [8] and [3], which have recently been confirmed by the results of experimental [9] and computational [10, 11] studies, only two modifications can be stable: the hexagonal form α - Al_2S_3 (SG $P6_1$, exists from room temperatures to ~1000 °C) and γ - Al_2S_3

phase with a trigonal structure of corundum (SG $R\bar{3}c$). The γ-phase exists at high temperatures: from ~1000 °C to the congruent melting point (~1070 °C). The α-modification, since it is the one that has a structure with stoichiometric vacancies, is of greatest interest. This structure is derived from the crystal lattice of "defective" wurtzite and was discussed in detail above in the analysis of "defective" gallium sesquisulfides of the α-Ga₂S₃ modification.

2.2. Physicochemical principles of using equilibria involving aluminum sulfides. Obtaining metallic aluminum using the carbosulfide method

Despite the apparent "academic nature" of the question of the existence of the AlS monosulfide on the phase diagram, which is formed only over a narrow temperature range and is not preserved upon hardening, the control of the processes of synthesis and decomposition of this substance seems very important for the development of a new scheme for the industrial production of metallic aluminum (see below). The search for new ways of obtaining metallic aluminum is connected with the fact that the traditional technology of smelting aluminum using electrolysis (the Hall-Héroult process) is accompanied by emissions of environmentally harmful carbon fluorides. These

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Table 1. Three-phase equilibria of condensed phases in the Al – S system according to from a number of literary sources. In the columns "Compositions" the numerical values are given for the phases listed in accordance with the corresponding equations of phase reactions. All calculated (according to different models) data of the table refer to the work [3], the experimental values – to the sources [6–8]

,		L 3/	•			. ,
Equation	Phase reaction type	T, °C	Compositions, mol. % S*			Source of information
		~ 660	-	0	60	Experiment, [6]
$L_1 = Al_s + \alpha - Al_2S_3$	Eutectic (degenerate)	660.5	0	0	60	Substitutional solution model (SSM)
		660.5	0	0	60	Associate model (AM)
		1060.0	_	-51	50	Experiment, [6]
$L_1 + L_2 = A1S$	Synthetic	1058.3	0	54.3	50	SSM
		1059.9	4.1	54.8	50	AM
		1040.0	56.0	50.0	~ 59	Experiment, [6]
$L_1 = AlS + \gamma - Al_2S_3$	Eutectic	1052.6	55.3	50.0	60	SSM
		1052.6	55.7	50.0	60	AM
	Catatectic	1010.0	50.0	•••	< 60	Experiment, [6]
$A1S = L_1 + \gamma - Al_2S_3$		1011.6	50.0	0	60	SSM
		1011.5	50	2.6	60	AM
		973.0	_	•••	•••	Experiment, [7]
$L_2 = \gamma - A1_2S_3 + L_3$	Monotectic	990.3	70.7	60	100	SSM
		970.6	63.6	60	100	AM
$L_3 = \alpha - Al_2S_3 + \beta - S$	Eutoptio	~ 115	~ 100	60	~ 100	Experiment, [8]
	Eutectic	115.2	100	60	100	SSM
	(degenerate)	115.2	100	60	100	AM
	Eutostoid	~ 95.5	~ 100	60	~ 100	1
$\beta - S = \alpha - Al_2 S_3 + \alpha - S$	Eutectoid	95.3	100	60	100	SSM
	(degenerate)	95.3	100	60	100	AM

fluorides (CF_4 with admixture of C_2F_6) are released as by-products at the sacrificial graphite anode, which is immersed in an $Al_2O_3 - Na_{3x}K_x[AlF_6]$ melt.

Step 1. Carbosulfidization of aluminum ore containing mostly Al₂O₂ [12]

One of the most promising alternative methods of obtaining Al is based on the so-called carbosulfide method, which consists of the following three stages. The Loutfy method is based on the reaction of aluminum ore (Al_2O_3) with coal (graphite) and sulfur vapor at a temperature of 1027 - 1227 °C with the formation of an Al_2S_3 melt (4):

$$Al_2O_{3(s)} + 3C_{(s)} + 1.5S_{2(g)} = Al_2S_{3(l)} + 3CO_{(g)}.$$
 (4)

A slightly different way of obtaining aluminum sesquisulfide was proposed by Sportel and Verstraten. According to this scheme γ -Al₂O₃ reacts with carbon disulfide vapors at a slightly lower temperature (\geq 750 °C), but under pressure from 5 to 35 atm:

$$Al_2O_{3(g)} + 3CS_{2(g)} = Al_2S_{3(l)} + 3CO_{(g)} + 1.5S_{2(g)}.$$
 (5)

Since CO and sulfur vapor react to form carbon monoxide sulfide, the latter equation is sometimes represented as:

$$Al_2O_{3(s)} + 3CS_{2(g)} = Al_2S_{3(l)} + 3COS_{(g)}.$$
 (6)

The carbon disulfide used in this scheme can be obtained directly in the same reactor by reacting natural gas with sulfur vapor:

$$CH_{4(g)} + S_{2(g)} = CS_{2(g)} + 2H_{2(g)}.$$
 (7)

Step 2. Incongruent sublimation of the Al_2S_3 melt. The resulting melt is heated under reduced pressure to temperatures from ~1330 to ~1630 °C. This results in the removal of some of the sulfur from the melt and the composition of the liquid (L₂) shifts to a composition with a sulfur content of 53–54 mol. %:

$$Al_2S_{3(l)} \to L_{2(\text{-AlS})} + S_{2(g)}.$$
 (8)

Step 3. Release of Al during the catathetic decay of AlS. Decrease in melt temperature of L_2 , obtained in the previous step, causes crystallization of the AlS phase (Fig. 1). Further cooling to $t < 1010\,^{\circ}\text{C}$ causes a disproportionation of aluminum monosulfide according to reaction (2). As a result, along with solid Al_2S_3 , almost pure liquid aluminum is formed. The molten metal flows into the bottom of the reactor and the aluminum sesquisulfide is redirected for reheating for the removal of the sulfur.

Recently, other applications of aluminum sesqui-chalcogenides have been discovered. In particular, α -Al₂S₂ was used in promising chemical power sources (CPS) [13, 14]. Al-S batteries operate at relatively low temperatures (~85 °C) due to the transfer of electrons from the anode (metallic Al or Al alloy) to the cathode - sulfur (S⁰). Free sulfur is intercalated into graphite-like boron nitride (α -BN) or distributed in an array of carbon nanotubes. When the battery discharges, sulfur turns into microcrystalline α -Al₂S₃. During charging, Al³⁺ ions are removed from the sulfide, and sulfur is converted to its original oxidized state S⁰. It should be emphasized that the possibility of fast charging of Al-S batteries is probably associated with the structural features of α -Al₂S₃, which contains *stoichiometric vacancies* $V_{\rm Al}$, which facilitates the diffusion of ${\rm Al^{3+}}$ ions in the solid phase. For the transfer of ${\rm Al^{3+}}$ ions in a liquid electrolyte, an ionic liquid is used in which the cations are represented by nitrogenorganic ions (for example, the cation of 1-ethyl-3-methylimidazolium), and the anions are [AlCl₄] and [Al₂Cl₇]⁻ ions. There are also reports on the high potential of such chemical power sources in which sulfur in the cathode material is partially or completely replaced by selenium [15], [16], and such CPS can be used even at room temperature.

2.3. The Al – Se system and aluminum selenide Al_2Se_3

In this part of the review, together with the Al – S system, we will also consider the corresponding selenium system. This is due to the fact that aluminum selenides have been less studied than sulfides, and the practical application of the only intermediate phase of this system, aluminum sesquiselenide (Al₂Se₃), is at the initial stage.

According to [3], a congruently melting narrowly homogeneous compound exists in this system at pressures close to atmospheric (Fig. 2), the monoclinic (Cc) structure of which is characterized by an ordered arrangement of vacancies and is considered in more detail in Chapter 2.1 (when considering the α' -Ga₂S₃ isostructural phase). At the same time, we note that the diagram presented in Fig. 2 is the result of calculations based on certain solution models and needs to be verified. The fact that this selenide was previously assigned the formula Al_zSe₄, may indicate a noticeable shift in the composition towards excess of aluminum relative to the ideal stoichiometry with a selenium content of 60 mol. % (compare with phase composition γ -Ga₂₊₈S₃ in the following described system).

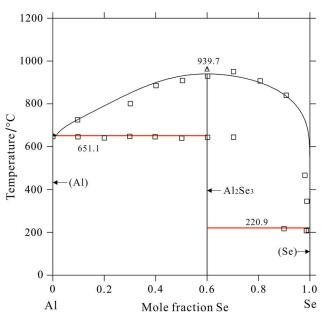


Fig. 2. *T-x*-diagram for the Al – Se system [3]. The results of the CALPHAD calculation using the substitution solution model are given together with the experimental data

3. The Ga - S system

When searching for ways of reproducibly synthesizing crystalline phases, especially in the case of the existence of many polymorphic modifications, it is highly desirable to find conditions under which these phases are stable relative to other compounds of this system*. In

^{*} Definitely, synthesis can be performed in a way that allows immediately obtaining a metastable or even unstable (under reaction conditions) compound. However this approach has the risk of non-reproducibility of the method.

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other words, when choosing such a strategy, the resulting modification must be present in the equilibrium phase diagram of the given system. In this case it is possible to obtain this phase under conditions close to equilibrium and then harden it. In our opinion, unsuccessful and mutually exclusive descriptions of the paths of formation of certain volumetric structures are associated with contradictory data on phase equilibria and on the phase diagram. The Ga – S system is an example of the need for a detailed study of the *P-T-x-* (or at least, *T-x-*) diagram.

Until now, the fact that this system contains solid congruently melting phases with the GaS and Ga₂S₃ stoichiometries, and also the fact that the latest stoichiometry corresponds to several modifications with "defective" crystal lattices, based on sphalerite or wurtzite structural types have not caused controversy [17], [18]. In addition, there was no question that the saturated vapor over gallium sulfides was represented by Ga₂S and S molecules. In this case, GaS sublimates incongruently with the formation of almost only Ga₂S, and for Ga₂S₃ in the homogeneity region there are congruently evaporating compositions [19, 20]. It should be noted that other chalcogenides of aluminum, gallium, and indium behave similarly during sublimation [1, 21].

However, until now, it has not been clear what other phases can be realized in this system and what is the structure of the corresponding diagram. In particular, in the studies of famous authors [20] and [19] GaS and Ga,S, sulfides were considered as the only crystalline intermediates in this system (Fig. 3). However, the study [22] provides a completely different pattern, where in addition to gallium mono- and sesquisulfide, subsulfide (Ga₂S), as well as Ga₄S₅, both with an incongruent melting type, were indicated. In relatively modern reference studies [23] and [24] the Ga – S system is also presented in accordance with [22] (Fig. 4). In addition, the existence of Ga₂S and Ga₄S₅ phases was confirmed in an independent study [25].

However, in our latest detailed studies [26-31] we obtained a *T-x-* diagram of the Ga-S system (Fig. 5), significantly different from either of the two images shown above. Moreover, the diagram in Fig. 5 turned out to be close to the images published in little-known Frenchlanguage studies [18, 32]. It should be emphasized that the ideas about the system and equilibria corresponding to Fig. 5 were obtained using various methods that confirm each other: X-ray diffraction analysis (XRD, HTXRD – up to 600 °C), *in situ* X-ray analysis using synchrotron X-ray radiation for high-temperature (up to ~

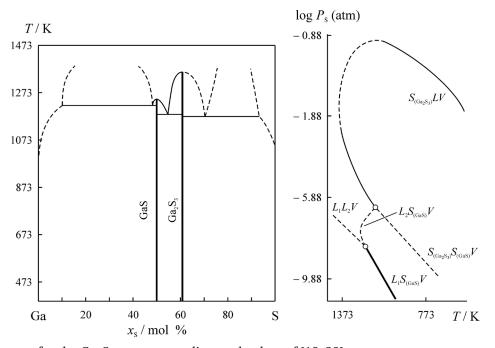


Fig. 3. T-x diagram for the Ga-S system according to the data of [19, 20]

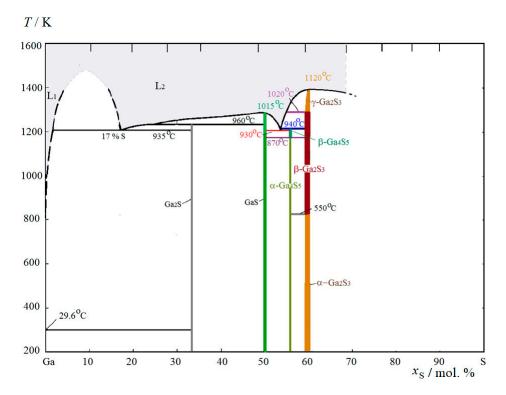


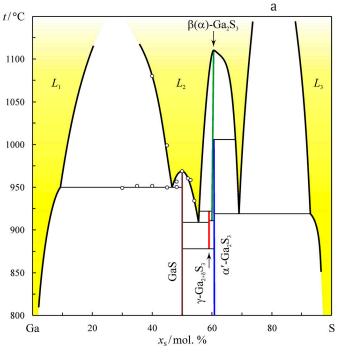
Fig. 4. *T-х* диаграмма системы Ga-S по данным [22]

1030 °C) studies of phase equilibria, differential thermal analysis (DTA) together with a new static method of thermal analysis [33], and methods of transmission electron microscopy (HRTEM, SAED). This diagram and the phases present in it are analyzed in detail in a recently published article [31]. We will not discuss this study in detail, just list the main features of the Ga – S system.

- 1. The existence of a Ga₂S solid phase was not confirmed experimentally. Moreover, there are no structural data on this substance in the literature. Most probably, the false ideas about solid gallium subsulfide arose from an analysis of the gross composition of the product of resublimation of gallium monosulfide from the hot zone to the cold zone. Indeed, the sublimation of GaS occurs incongruently with the predominant formation of Ga₂S vapor. However, upon condensation, the subsulfide vapors disproportionate, yielding a sample of Ga + GaS heterogeneous powder.
- 2. For the Ga_4S_5 phase found in [22], most likely, an error in the determination of the composition was made. Probably, in this study the authors obtained one of the phases of the Ga_2S family, namely, γ - $Ga_{2+\delta}S_3$ with a sulfur content of about 59.3 mol. %.
- 3. If gallium monosulfide is a practically completely stoichiometric compound, for gallium sesquisulfide there is a whole family of phases with "defective" structures of sphalerite or wurtzite. In this case, the phase with the highest ordering of stoichiometric vacancies during genesis from the wurtzite α' - Ga₂S₃ structure is, according to the ideas [18], the only phase that is stable at room temperature. Moreover, like GaS, it is narrowly homogeneous and almost completely stoichiometric. Other phases of this family can be stable (relative to other phases) only at high temperatures. In this case, it is mostly shifted towards gallium relative to the ideal composition – up to 0.7 mol. % – "defective-sphalerite" modification γ -Ga₂₊₈S₃, which is realized on the phase diagram in a narrow temperature region of about 900 °C. Higher temperature forms – α - and β -Ga₂S₃ are grouped closer to the composition of 60 mol. % S. All of them can be isolated individually and stored at room temperature after annealing under equilibrium conditions, followed by quenching.
- 4. Obtaining one or another modification of the Ga₂Se₃ family, probably, is very dependent on the presence of even small concentrations of impurities, as evidenced by the following facts.

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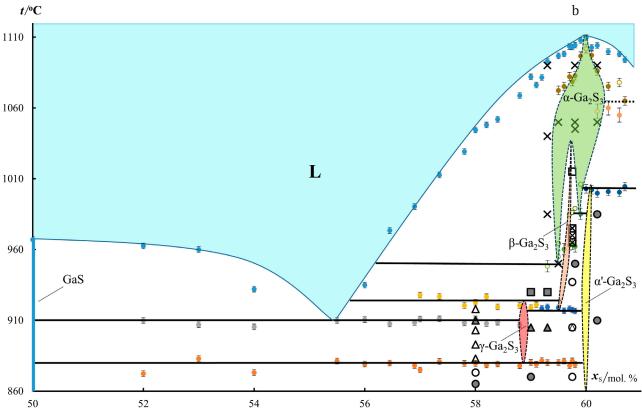


Fig. 5. New data on the T-x diagram of Ga – S: general view [29] (a) and a detailed fragment in which 4 "defective" modifications of the Ga₂S_x family are realized [31] (b)

a). In particular, we observed stabilization of the cubic γ -Ga₂₊₈S₃ phase when introducing already 0.5 mol. % iron [34]. Hexagonal modifications, α - and β -Ga₂S₃, are stabilized

by the addition of manganese [35]. These modifications were observed at temperatures much lower than those allowed for this system by the phase diagram of the Ga – S system,

even taking into account a small possible supercooling.

b). The authors of [36] reported the isolation of a cubic form during crystallization from potassium iodide or lead chloride melts [36]. Moreover, in the latest study, the order of releasing of equilibrium phases of the Ga_2S_3 family at different temperatures turned out to be completely different than it follows from T-x-diagrams of the Ga – S system (compare Fig. 5 and Fig. 6).

c). A cubic γ -phase was also released during the crystallization of glasses of the Ga-Ge-S system (the Ga₂S₃ – GeS₂ section) at 750 °C. It should be noted that for pure gallium sesquisulfide the γ -Ga_{2+ δ}S₃ phase becomes metastable at temperatures below 878 °C and quickly turns into α' -Ga₂S₃ at t < 840 °C (with slow cooling or annealing) [37].

d). In the production of gallium sesquisulfide by the oxidation of molten gallium triiodide with elemental sulfur at 500 °C the γ -Ga₂₊₈S₃ cubic high-temperature modification was also released; with a slight increase in temperature (up to 650 °C) another phase with a hexagonal structure was released ($P6_1$) [38]. It should be noted that for *pure* gallium sesquisulfide both modifications should at the specified temperatures unambiguously transform into monoclinic α' -Ga₂S₃*.

Compounds from the gallium sesquisulfide family should be very promising for materials science (luminescent properties [39], [40], nonlinear optics [40], etc.). However, the incomplete reproducibility of the isolation of each individual modification significantly limits their use. We hope that a refined phase diagram of the Ga–S system, as well as detailed studies of the ternary diagrams Ga–S–X (X is an impurity), especially taking into account the extremely high sensitivity of the stability of various modifications to very small concentrations of impurities will contribute to the solution of the question of the use of phases of the Ga₂S₃ family in modern materials science.

4. The Ga - Se system

Currently, the Ga–Se system appears to be the poorest among A^{III} – B^{VI} based on the number of intermediate solid phases [3, 21]. The *T-x*-diagram of the discussed system contains two polymorphic modifications of layered gallium monoselenide [40] and two modifications of gallium sesquiselenide [41], [42] (Fig. 7).

"Defective" compounds of the Ga_2Se_3 family are represented by a high-temperature β - Ga_2Se_3 modification with a sphalerite structure (with disordered vacancies) and a low-temperature

and the designation β -Ga₂S₃ was assigned to the wurtz-ite-like "defective" form with SG $P6_3mc$.

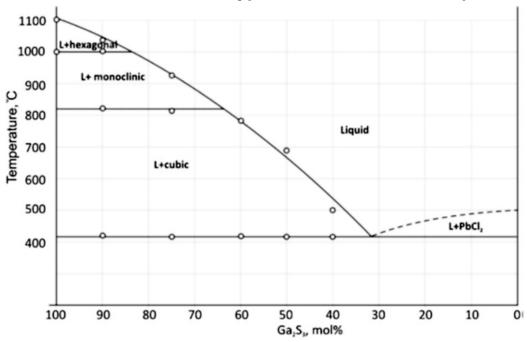


Fig. 6. *T-x* diagram for the Ga₂S₃ – PbCl₂ according to the data of [36]

^{*} The authors [38] for the hexagonal modification with SG $P6_1$ used the name β -Ga₂S₃. In the majority of literary sources and in our studies this phase appears as α -Ga₂S₃,

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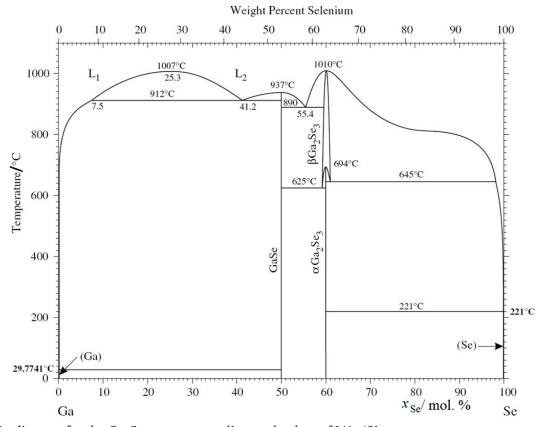


Fig. 7. T-x diagram for the Ga-Se system according to the data of [41, 42]

monoclinic α -Ga₂Se₃ form, which is isostructural to α' -Ga₂S₃ (with ordered vacancies) [41, 42]. At the same time, it should be noted that in our preliminary studies we have not yet been able to confirm the equilibrium existence of high-temperature α -phases, despite a fairly detailed study. In the latter, in addition to DTA, *in situ* X-ray powder analysis with X-ray synchrotron radiation was used. This allowed the Ga–Se system sample to be annealed under specific high-temperature conditions for several hours to establish equilibrium.

4.1. Regulation of the GaSe polytypic structure and composition by the SSCTR method

In our study, we were able to reproducibly obtain individual polytypes of layered gallium monoselenide [43–45]. For this purpose, a nonisothermal version of phase composition control (both phase and within the homogeneity region) was used using the method of selective chemical transport reactions (SSCTR).

The essence of the SSCTR is the introduction or removal of one of the components of the

solid phase (in our case, Ga) during a selective gas transport reaction. In an experiment on regulating the phase composition using the SSCTR method, the initial sample is placed at one end of the ampoule, and the batch, which is the source of the transported component, is placed at the other end. Also, a certain amount of transport agent is introduced into the ampoule, which, under experimental conditions, produces unsaturated vapor. Then, after removing the air, the sealed ampoule is placed in an oven in a way that the higher temperature (T_{γ}) corresponds to the sample, and the lower temperature (T_1) corresponds to the batch. In the system, selective transfer of only one component between the batch and the sample occurs. It should be emphasized that this transfer gradually ceases as the controlled sample reaches a certain phase composition (i.e. a certain structure) and a certain non-stoichiometry (i.e. a specific composition within the homogeneity region). As a result, a steady state without mass transfer is realized in the system, in which the sample can exist for an indefinite period of time at the fixed temperatures of the hot and cold zones (T_2, T_1) [46].

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In our studies [43], [46], to regulate the composition and structure of gallium selenides, pure gallium was used as a batch and gallium triiodide was used as a transport agent. The chemistry of selective transfer was reduced to a reversible reaction (9):

$$2\,Ga^{\text{in the composition of Ga selenides}} + \,GaI_3^{\text{vapor}} \rightleftarrows 3\,GaI^{\text{vapor}} \,. \quad (9)$$

It should be noted that at temperatures up to $\sim 900\,^{\circ}\text{C}$ gallium selenides have only small values of saturated vapor (Ga₂Se, Se₂ pressure does not exceed several mm Hg [1, 21]). This fact, as well as the absence of stable forms of binary and ternary selenium iodides in the pair, lead to the fact that the role of selenium or its compounds in mass transfer is minimal. The latter allows the selective nature of iodide chemical mass transfer of gallium selenides.

The results of the study showed that γ -modification of GaSe (alternating packets of the ...ABCABC...type, Fig. 8) exists on the phase diagram only with an excess of selenium and only at high temperatures close to the melting point. The ϵ -GaSe polytype (alternating packets of the ... ABAB-...type, Fig. 8) always turns out to be stable relative to other phases at low temperatures

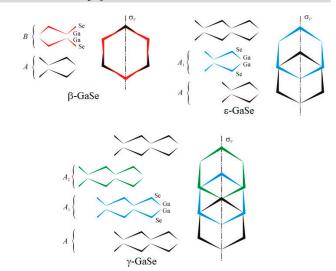


Fig. 8. Structures of some polytypes of GaSe (according to [44])

(up to \sim 700 °C), as well as at temperatures up to melting in the case of high gallium content within the homogeneity region [43, 44]. On the *T-x*-diagram these polytypes according to our research data are shown in Fig. 9.

Thus, it is possible to solve a rather complex problem of reproducible obtaining of a certain polytype. The deformation nature of the

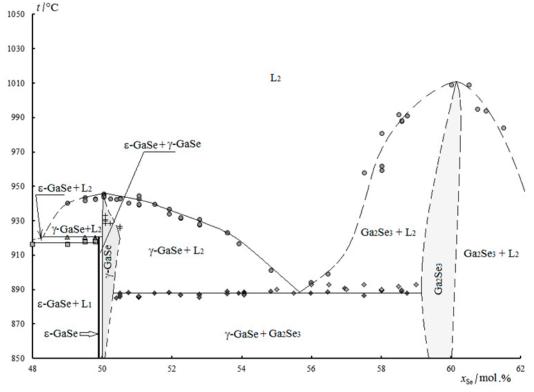


Fig. 9. High-temperature region of *T-x*-diagram for the Ga – Se system according to [45]

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transformation of one polytype into another allows in this case to preserve single crystals of substances without their destruction. It also should be noticed that in [43], [44] the possibility of regulating not only the polytypic structure, but also the composition within the homogeneity regions of these phases is shown.

5. The In - S system

When moving to indium chalcogenides, a much greater variety of intermediate phases is observed. This is usually associated with the greater tendency of indium to form compounds with a larger proportion of the ionic component of the chemical bond (especially compared to gallium) with a variety of charge states (In⁺¹, In^{+3} , to a lesser extent In^{+2}). Thus, on the In - Sphase diagram ([47-49], Fig. 10) there are InS phases (two modifications: the low-temperature form InS, which transforms via a distectoid path into the high-temperature form), In_6S_7 , $In_{3-7}S_4$ (according to a number of sources, it is also α -In₂S₂), as well as polymorphic modifications of indium sesquisulfide: low-temperature β -In₂S₂ and high temperature γ -In₂S₂. The high temperature InS form, as well as n_6S_7 and $In_{3x}S_4$ melt incongruently. The γ - In_2S_3 is the most resistant to high temperatures and melts congruently at 1097 °C [49] (1090 °C according to [50]).

Taking into account the relatively low values of partial pressures over indium sulfides, their incongruent sublimation and the difficult kinetics of establishing equilibria between condensed phases and their own vapor, it is difficult to use classical tensimetric research methods for the In–S system, as well as classical methods for regulating the composition of these sulfides through the vapor phase. For this reason, the auxiliary component (AC) method seems to be promising both for a detailed analysis of the considered system and for tasks on the synthesis of indium sulfides.

The In-S system was studied by the AC method, primarily using hydrogen with the aid of a spectrophotometric technique [47], [48], [51]. The studied equilibrium in this case can be written as (10):

$$S^{\text{in the composition of In sulfides}} + H_2^{\text{gas}} \rightleftharpoons H_2 S^{\text{vapor}}$$
. (10)

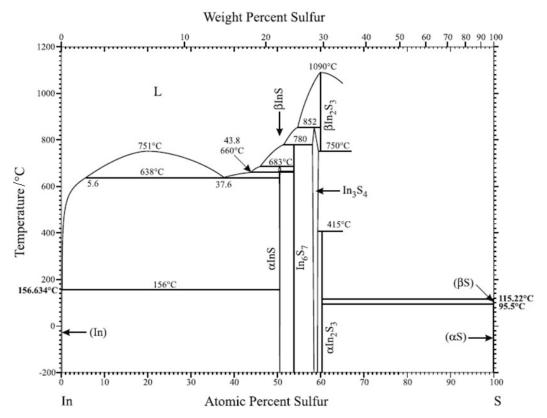


Fig. 10. T-x diagram for the In – S [50] system, based on the data of [49]

To achieve equilibrium, the studied sample of the In – S system was kept in contact with hydrogen at a temperature of 400 to 725 °C. After such annealing for many hours, the reactor with the substances (a sealed quartz cuvette) was quenched and the absorption spectrum in the region of hydrogen sulfide absorption was studied ().

In addition to the previously established set of intermediate phases of the In–S system (InS, In $_6$ S $_7$, In $_{3-x}$ S $_4$, In $_2$ S $_3$ modifications) it was established that the In $_{3-x}$ S $_4$ * phase has a wide homogeneity range: from ~ 58.0 to 59.8 mol. % S. In this case, the behavior of the concentration dependences of the chemical potential of sulfur** in this area of homogeneity turned out to be very unusual. These dependences consisted of alternating almost horizontal and inclined sections, which, according to the authors [51], may indicate a tendency of this cubic phase to disintegrate into separate independent modifications with similar but different compositions.

The auxiliary component method was also used to solve problems related to the synthesis of substances with a given composition, both phase and within homogeneity regions. In these experiments, the goal was to specifically vary the composition of pre-synthesized indium sulfides, both fine-crystalline films and bulk samples [52-54]. For this purpose, chlorine in the form of indium chlorides was used as an auxiliary component (also known as a transport agent). In this case, transfer via a reversible reaction (11) was used, similar to (9):

$$2In^{in \ the \ composition \ of \ In \ sulfides} + InCl_3^{vapor} \rightleftarrows 3InCl^{vapor}$$
 .
(11)

In this case, the sulfur atoms chemically bound in indium sulfide practically do not pass into vapor, which allows us to discuss the selectivity (11). However, the use of metallic

T = const).

indium as a control charge in such experiments turned out to be impossible because indium produces much more stable In⁺¹ halides compared to similar gallium compounds. As a result, equilibrium (11) turns out to be extremely shifted to the right in the equilibria of the InCl₃ vapor with In_{met.} To circumvent this difficulty, we used the dilution of indium in the batch with a chemically inert diluent. This allowed to reduce the activity of indium towards higher halides. Indeed, when using In–Au melt solutions, it was possible to change the direction of indium mass transfer between the sample and the batch and reproducibly obtain a sulfide of the required composition and structure depending on the gold content in the batch [44, 53]. In [54] the possibility of regulating the composition of indium sulfides using a batch of heterogeneous mixtures of indium sulfides (for example, $In_6S_7In_{3-x}S_4$) instead of expensive gold was demonstrated.

6. The In - Se system

According to our latest studies [30, 55] and data of other authors [56, 57], the T-x diagram of the In – Se system is distinguished by a huge variety of phases even in comparison with the In – S system. Solid selenides are realized in a small concentration range (40–60 mol. % S) and demonstrate pronounced polymorphism (Fig. 11). These are In₄Se₃ selenides (however, there is no similar sulfide, or gallium chalcogenides with such a stoichiometry), InSe (a layered structure similar to ε -GaSe), In₆S₇ and In₂Se₃ (several modifications). In our latest studies, which were carried out using a new spectrophotometric technique with an auxiliary component ([58-60]), the following was discovered [30], [56].

- 1. It was shown that high temperature (650–890 °C) modification of δ -In₂Se₃, is wurtzite-like (SG $P6_3mc$; may contain stoichiometric vacancies), while in the literature it is classified as SG $P\bar{3}m1$.
- 2. Near the composition corresponding to the stoichiometry of In_6Se_7 , there are two phases with similar but different structures and compositions. When this was revealed, alloys of the In Se system with a selenium content of 54.0 and 57.5 mol. % Se were studied using X-ray phase analysis. Before the study, the alloys were homogenized in repeated long-term annealing

^{*} The indium mono- and sesquisulfides differ from the structures of other $A^{III}B^{VI}$ mono- and sesquichalcogenide compounds (except γ -In $_2S_3$). In particular, the In $_3$ - $_3S_4$ phase (often not quite correctly called β -In $_2S_3$) has the structure of a "defective" inverted spinel (see Part 1 of this review).

^{**} More precisely, we are discussing the value (let's denote it $\ln(K_p^*)$, see [51]), which at a constant temperature T is linearly related to chemical potential $\mu_{\rm S}$ in the solid phase [51]: $\ln(K_p^*) = \frac{\mu_{\rm S}}{RT} + Y$, where Y is a constant value (at

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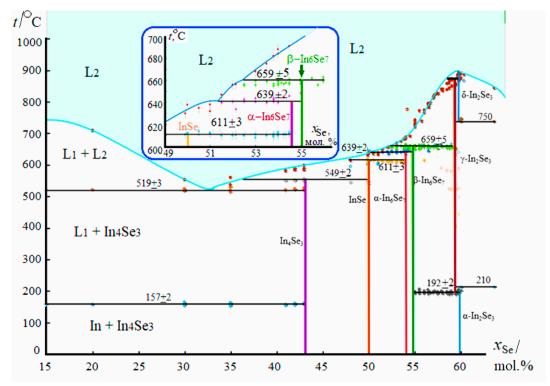


Fig. 11. T-x-diagram of the In – Se system in the composition range of 15.0 – 63.4 mol. % Se Se [30]

procedures (500 °C, 72 h) and grinding the powder (2 cycles in total). It was found that the first sample was represented by a monoclinic (SG $P2_{\perp}$) structure, and the second turned out to be a mixture of two phases and contained α -3R-In₂Se₂ as an impurity. The main phase of the sample with an increased selenium content (57.5 mol. %) compared to the stoichiometry of In₆Se₇ was identified as monoclinic and resembled the In Se. structure described above. However, it was well described by *another* space group (P2, m) and, therefore, was a separate phase. Both structures $(P2_1 \text{ And } P2_1/m)$ are mentioned in scientific papers, however, for the description of the In₆Se₇ structure [61-64] the second SG is used more often [61]. It is assumed that the crystal lattice of the compound with SG $P2_1/m$ corresponds to the phase with the stoichiometry In₉Se₁₁ described in a number of studies [62, 63], but still not structurally characterized. The composition of this phase, which we designate as β -In₂Se₇ [30], was close to 55.0 mol. % Se. The composition of the modification with SG P2, is richer in indium (we denote it as β -In₆Se₇ [30]) better corresponds to the stoichiometry of In Se, (~54 mol. %Se).

3. In our study of this diagram, we did not find any structures that could correspond to the

stoichiometries of In₅Se₇ and In₃Se₄. Phases with such compositions have not been confirmed either by X-ray powder diffraction studies or by thermographic or tensiometric experiments.

Thus, the main compounds of indium and selenium are $In_4Se_3(P_{nnm})$, $InSe\ (P6_3/mmc)$, In_6Se_7 (two similar modifications, but differing in composition: $P2_1$ And $P2_1/m$) and In_2Se_3 (represented by at least three modifications, $3R-\alpha-In_2Se_3$, $\gamma-In_2Se_3$, as well as $\delta-In_2Se_3$). Most of these phases decompose upon heating by peritectic reactions (except for congruently melting $\delta-In_2Se_3$). Over the course of thermal analysis (DTA, chemiluminescence analysis [33]), we obtained the following quantitative data on the temperature stability of condensed phases of the In-Se system and equilibria involving these phases (Table 2).

Out of the issues that are not yet completely clear, we note that this system requires additional studies of samples with compositions close to 60 mol. % selenium content. In these studies, it is still necessary to determine in more detail the positions of the modifications of the In_2Se_3 family on the corresponding phase diagram. Today, only the exact position of the low-temperature $3R-\alpha-In_2Se_3$ phase does not raise questions.

Table 2. Observed effects and phase equilibria in the In – Se system

No	Number of exp. points	Temperature at which the effect was observed, <i>t</i> , °C	The compound or range of compounds for which the effect was observed x_{s_e} , mol. %	Corresponding phase equilibrium effect
1	9	156.5±2	30.0-43.01	$In_4Se_3 + In \Leftrightarrow L$, eutectic
2	9	519±2	30.0-43.01	$L_1 + In_4Se_3 \Leftrightarrow L_2,$ monotectic
3	7	549±3.7	41.0–49.9	In₄Se₃ ⇔ InSe + L, peritectic
4	25	611±3.3	49.9-55.0(54.5)	InSe $\Leftrightarrow \alpha$ -In ₆ Se ₇ + L, peritectic
5	15	639±2.0	51.5–55.0	α -In ₆ Se ₇ \Leftrightarrow β-In ₆ Se ₇ +L, peritectic
6	30	659±4.5	53.5–59.5	β -In ₆ Se ₇ \Leftrightarrow γ-In ₂ Se ₃ +L, peritectic
8	18	191±2.0	55.0-60.0	β -In ₆ Se ₇ + α -In ₂ Se ₃ \Leftrightarrow γ -In ₂ Se ₃ , eutectoid
9	2	890±5.0	60.0	σ -In $_2$ Se $_3 \Leftrightarrow L$, congr. melting
10	52	от ~520 до ~891	15.0-63.35	$\begin{array}{c} S \Leftrightarrow L \\ (liquidus) \end{array}$
11	3	211±2.0	60.0-63.5	α -In ₂ Se ₃ \Leftrightarrow γ-In ₂ Se ₃₊ L, peritectic

7. Conclusions

In conclusion of our review, we will summarize the most important results of the study. First of all, let's analyze, *what do we know* about $A_1^{III}-B_1^{VI}$ systems, as well as phases of families $A_1^{III}B_1^{VI}$ and $A_2^{III}B_3^{VI}$, to which this study is mainly devoted.

1. In all the discussed systems, non-molecular crystalline compounds with $A_1^{\rm III}B_1^{\rm VI}$ and $A_2^{\rm III}B_3^{\rm VI}$ stoichiometries are realized. A partial exception is aluminum monoselenide, but, as we already noted, this compound is also realized in the form of films.

2. A₁^{III}B₁^{VI} **mono**chalcogenides form typical layered structures that have significant similarities with graphite (analogues of graphene, nanotubes and other nanoformations, intercalation reactions, including those during which there is partial damage to the structure of the intercalated crystal itself). Among the monosulfides and monoselenides of indium and gallium, the only partial exception is indium monosulfide, in which, despite the four-layer structure, the bonds between the individual

layers can no longer be exclusively classified as van der Waals.

3. **Sesqui-chalcogenides** of aluminum, gallium and indium provide families of "defective" semiconductors, in which crystalline compounds with ordered and disordered stoichiometric vacancies based on wurtzite and sphalerite structures, containing stoichiometric vacancies as structural units, are realized. A partial exception is the In–S system: of the three modifications, close in stoichiometry to indium sesquisulfide, only the highest-temperature one can be classified as a "diamond-like defective" structure based on sphalerite or wurtzite. The other two forms, despite the fact that they also contain stoichiometric vacancies, are derivatives of defective spinel.

4. All intermediate solid compounds in $A^{\text{III}} - B^{\text{VI}}$ systems are characterized by kinetic difficulties and "sticking" in metastable states. This is especially important when working with thermal analysis methods. In particular, in DTA experiments we observed the effect of "switching"

from one type of phase diagram to another when the heating rate threshold was exceeded by more than 13 K/min [49]. Sufficiently reproducible and clearly interpretable results appear only at the lowest heating rates (0.5–2 K/min). The static method of thermal analysis was developed for the elimination of this problem [33]. In addition, in order to circumvent the difficulties with kinetic difficulties, the auxiliary component method can be recommended [54], which can be used both in solving research problems and for the synthesis of substances with precisely specified compositions [46].

5. Phase equilibria in A^{III}–B^{VI} systems, especially with the participation of "defective" phases of families are very sensitive even to small concentrations of impurities. Doping with even small concentrations of impurities can fundamentally change the pattern of phase equilibria and the appearance of phase diagrams of systems, which can no longer be considered binary, but must be considered as multi- (at least three-) component.

Let us now turn to the problems that remain open to the discussed systems and phases of $A_1^{III}B_1^{VI}$ and $A_2^{III}B_3^{VI}$ families.

- 1. Now, the relationship between different defects in phases can be considered as an open problem. In particular, *a priori* it is assumed that in the γ -Ga₂₊₈S₃ cubic modification a noticeable phase deviation from ideal stoichiometry (59.3 instead of 60.0 mol. % S) occurs due to the "healing" of part of the stoichiometric vacancies by gallium atoms. However, this assumption has not yet been verified by anyone.
- 2. Another issue is related to finding a more adequate definition of what stoichiometric vacancies are. The point is that in some cases of ordering (for example, for α' -Ga₂S₃), it is no longer possible to discuss such vacancies, since a new type of crystal lattice is formed, in which the stoichiometry corresponds to a new structural type. At the same time, many properties (the possibility of creating interlayers during epitaxy, high diffusion coefficients, preserved radiation resistance, etc.) indicate the presence of such vacancies as independent structural units in crystal lattices.
- 3. For the monochalcogenides there is still no understanding of when, under what conditions

and by what reagents intercalation interactions can be carried out, and when such reactions are impossible. This also includes the question of which areas are affected by intercalation if it occurs and which areas are not affected (and for what reason). There is currently no understanding of the question of how intercalation can be carried out, achieving equilibrium in a way, that this equilibrium can be controlled by changing the concentration of the guest substance in the layered crystal and its ordered or disordered arrangement in the van der Waals gaps.

These questions await answers in future studies.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Thermodynamic study of zinc antimonides by the electromotive force measurements

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Abstract

Zinc antimonides and phases based on them are of great interest as earth-abundant, low-cost, and environmentally friendly thermoelectric materials. The present work demonstrates the results of a thermodynamic study of the ZnSb and $\mathrm{Zn_4Sb_3}$ compounds by a low-temperature electromotive force (emf) method with a glycerol electrolyte in the 300-430 K temperatures range.

Measurements were performed using equilibrium samples from the ZnSb+Sb and $ZnSb+Zn_4Sb_3$ two-phase regions of the Zn-Sb binary system. The phase compositions of prepared samples were controlled by means of the powder X-ray diffraction (PXRD) method. Using the least square method, the linear equations of temperature dependences of the emf data were obtained.

Based on these equations and relevant thermodynamic expressions, the partial molar Gibbs free energy, enthalpy, and entropy of zinc in alloys were calculated. Utilizing the phase diagram of the Zn-Sb system, the virtual-cell reactions for both binary compounds were determined, based on which their standard thermodynamic functions of formation and standard entropies were calculated. A comparative analysis of the obtained results with available literature data was carried out.

The results of the current work are highly accurate and can be considered a new contribution to the thermodynamics of zinc antimonides.

Keywords: ZnSb, Zn, Sb, emf method, Electrochemical cells, Glycerol electrolyte, Thermodynamic functions

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1. Introduction

Metal pnictides are a class of compounds which are intriguing due to their promising electronic, optical, magnetic, etc. properties, that have resulted in a wide range of applications in various fields of modern technology and industry. So that binary and more complex pnictides and phases based on them are considered promising materials for applications such as high-efficiency light-emitting diodes, solar cells, quantum transport, low-power high-speed electronics, mid-to long infrared wavelength optoelectronics, thermoelectric (TE) generation and so on [1– 4]. Some of them have gained recognition in physics and materials science as 3D topological Dirac semimetals and have been suggested for use as Weyl semimetals, axion insulators, and topological superconductors, making them a unique parent compound for the study of these states and the phase transitions between them [5-8].

Among binary pnictides, Zn-Sb phases are the most investigated materials showing promising TE performance. In addition, they are earthabundant, low-cost, environmentally friendly, and stable at high temperatures up to ~1000 K materials which make them one of the best choices for the TE industry since the discovery of the Seebeck effect [9–12].

Although zinc antimonides have glass-like thermal conductivity, which makes them attractive for TE applications, their poor carrier concentration compared with state-of-the-art TE materials [13] has inspired many scientists to try to improve their electrical properties by adding dopants such as In, Al, Sn, etc. [14–16].

The phase diagram of the Zn-Sb system has been extensively studied experimentally and theoretically by many authors [17–21]. The phase diagrams reported in these studies are very similar, although the topology is slightly different. The true equilibrium phase diagram determined by a careful study considering the differences among all the previous studies was reported by H. Okamoto [22]. According to [22], Zn-Sb phase diagram is characterized by the formation of 3 antimonides: ZnSb, Zn₄Sb₃, and Zn₃Sb₂. The first compound melts at 818 K by peritectic decomposition and is stable below its melting temperature. Zn₄Sb₃ and Zn₃Sb₂ compounds

have distectic melting at accordingly 837 and 838 K temperatures. Zn_4Sb_3 has 3 modifications: stable below 768 K (α - Zn_4Sb_3), between 768 and 803 K (β - Zn_4Sb_3) and above 803 K (γ - Zn_4Sb_3), respectively. Zn_3Sb_2 appears in low- and high-temperature modifications and is metastable at room temperature [22].

In a recent study, an updated phase diagram of the system Zn–Sb provides new details about the composition, transformation, and stability temperatures for the four binary phases [23]. According to the results of [23], the Zn–Sb system contains two well-known thermoelectric materials: Zn_{1–8}Sb and Zn_{13–8}Sb₁₀ ("Zn₄Sb₃"), and two other phases: Zn_{9–8}Sb₇ and Zn_{3–8}Sb₂, which are only stable at high temperatures. The chemical formula of the well-known Zn_{4–8}Sb₃ phase was suggested as Zn_{13–8}Sb₁₀ according to its structure and composition.

Thermodynamic properties of compounds are essential for forecasting the thermal behaviour and microstructure evolution of alloys as temperature changes. As scientists explore zinc antimonides for potential applications in electronic and optoelectronic devices, a precise comprehension of their thermodynamic properties becomes imperative, paving the way for optimized performance and leveraging their capabilities in technological advancements. The thermodynamic properties of zinc antimonides were studied by different authors and summarized in modern review papers and databases [24-27]. These works are mainly dedicated to the determination of their enthalpies and Gibbs free energies of formation with calorimetric and high-temperature emf methods, as well as by thermodynamic assessment. Later, in the discussions section of the paper, the results of these studies will be discussed, focusing on experimental methods and models that have been used to estimate related thermodynamic properties and will be compared with the ones obtained in the current study. Therefore, to obtain a more reliable set of thermodynamic functions, it is recommended to use experimental data obtained under conditions as close as possible to the standard.

Various modifications of the *emf* method are commonly employed for the thermodynamic analysis of binary and more complex inorganic

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systems. Depending on different factors such as measurement temperature, the composition of compounds, etc., different types of liquid and solid electrolytes can be applied for *emf* measurements [28-32].

Consequently, the present paper is dedicated to the thermodynamic study of zinc antimonides by low-temperature *emf* measurements.

2. Experimental part

The compositions of the prepared samples, as well as their synthesis and thermal annealing conditions were selected with respect to the phase diagram of the Zn-Sb system [22]. Alloys of different compositions from the ZnSb + Sb and ZnSb + Zn₄Sb₇ two-phase regions have been selected to study the thermodynamic properties of zinc antimonides stable at room temperature. Two samples with 52 and 70 at.% Sb composition from the ZnSb + Sb phase region and two samples with 43 and 47 at.% Sb composition from the ZnSb + Zn₄Sb₇ two-phase region of the system were selected for the thermodynamic study of the ZnSb and Zn₄Sb₂ compounds, respectively. Highpurity elements (Zinc lumps - Sigma-Aldrich, CAS number 7440-66-6 and Antimony pieces -Alfa-Aesar, CAS number 7440-36-0) were used for sample preparation.

For synthesis, calculated amounts of zinc and antimony were weighed in an analytical balance and inserted into the quartz ampoules. Ampoules were evacuated up to $\sim 10^{-2}$ Pa pressure and heated up to 900 K temperature. After 2-3 hours synthesis process, the temperature was gradually decreased to 600 K and the samples were kept at this temperature for ~ 500 hours.

The PXRD technique was used to confirm the phase composition of the prepared equilibrium samples. Data collection was performed at room temperature using the D2 Phaser diffractometer with $CuK\alpha$ emission. Samples were scanned from ~ 5 to 75°. The Topas 4.2 profile modelling software was used to examine and analyse the recorder diffraction patterns.

PXRD spectrum of one sample from each phase region is given in Fig. 1. As can be seen, the PXRD spectrum of the sample with 47 at.% Sb composition (Fig. 1a) is composed of diffraction peaks of the ZnSb and Zn₄Sb₃ compounds, while the powder diffractogram of the sample with

70 at.% Sb composition (Fig. 1b) consists of diffraction lines of the ZnSb compound and an elemental antimony. Any signals belonging to other phases have not been detected in the diffraction patterns of samples, which confirms the complete synthesis and homogenization processes.

To start *emf* measurements, the following electrochemical cell was constructed:

(-) Zn(s) / glycerol + KCl + ZnCl₂ / Zn-Sb alloy (s) (+) (1)

The most active element of the system, namely zinc was used as the left electrode, while prepared equilibrium alloys of the Zn-Sb system were used as the right electrodes. To prepare the right electrodes, previously synthesized and annealed samples were powdered, pressed into tablets of 5-6 mm in diameter, and attached to the molybdenum rods. Molybdenum rods and electrodes were covered by glass coatings to prevent possible contact between them inside an electrolyte solution.

A liquid electrolyte of the constructed cell was a glycerol solution of KCl (Sigma Aldrich, 99.999%) with a small addition of anhydrous ZnCl₂ (Sigma Aldrich, 99.999%). Due to the presence of moisture and oxygen in the electrolyte, glycerol was thoroughly dehydrated and degassed at ~350 K under a dynamic vacuum. Preparation of the electrolyte and electrodes, as well as the assembly technique of the electrochemical cell (1) were put into practice as described in [33, 34].

A Keithley 2100 6 $\frac{1}{2}$ The digital multimeter, having 1014 Ω input resistance and ± 0.1 mV accuracy, was used to collect *emf* data in the 300- 430 K temperature interval. A chromelalumel thermocouple and a mercury thermometer were used for temperature measurements. After keeping the cell at ~350 K for 40–60 hours, the first equilibrium values of the potential difference were obtained. Then subsequent measurements were taken every 3-4 h when a particular temperature was established. For equilibrium values of the *emf*, the difference between repeated measurements was not higher than 0.2 mV at the established temperature.

3. Results and discussions

According to experimental measurements' results, the *emf* values were constant within each

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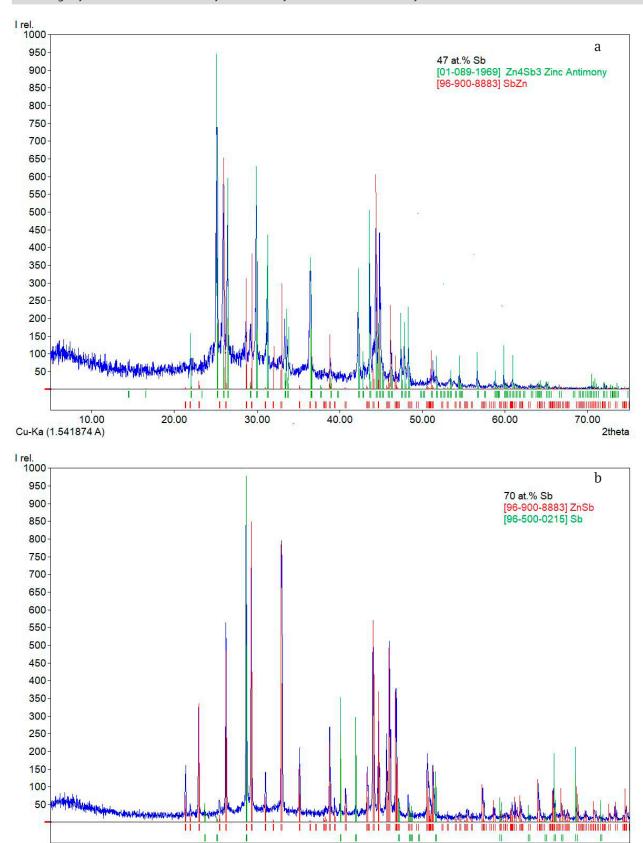


Fig. 1. PXRD spectrums of samples with a) 47 at. % Sb and b) 70 at. % Sb composition along the Zn-Sb binary system

40.00

50.00

60.00

30.00

10.00

Cu-Ka (1.541874 A)

20.00

70.00

of the ZnSb + Sb and $Zn_4Sb_3 + ZnSb$ two-phase fields, independent of the overall composition of electrode alloys. Temperature dependencies of the *emf* values for both samples are illustrated in Fig. 2. Analysis of the $E \sim f(T)$ dependences of the alloys has shown that they are practically linear (Fig. 2). It confirms the stability of the compositions of coexisting phases in above mentioned heterogeneous phase areas in the temperature range under study and gives a base for estimations of the partial entropy and enthalpy from values of the temperature coefficients of the *emf* [35, 36].

Collected experimental data were processed using least squares fitting with the Microsoft Office Excel computer program to give linear equations

$$E = a + bT \pm t \left[\frac{\delta_E^2}{n} + \delta_b^2 (T - \overline{T})^2 \right]^{1/2},$$
 (2)

where a and b are constant coefficients, n is the number of pairs of experimental E and T values; δ_E^2 and δ_b^2 are the error variances of the emf readings and b coefficient, respectively; \overline{T} is the average of the absolute temperature; t is Student's test. With the number of experimental points n=30, and the confidence level equal to 95%, the student's test is $t \le 2$. Obtained linear equations of the type (2) are presented in Table 1. Based on these linear equations and using the thermodynamic expressions below given, the partial molar Gibbs free energy, enthalpy, and entropy of zinc in alloys have been calculated [33–36]:

$$\Delta \overline{G}_{Zn} = -zFE, \qquad (3)$$

$$\Delta \overline{S}_{\rm Zn} = zF \left(\frac{\partial E}{\partial T} \right)_P = zFb , \qquad (4)$$

$$\Delta \overline{H}_{\rm Zn} = -zF \left[E - T \left(\frac{\partial E}{\partial T} \right)_P \right] = -zFa. \tag{5}$$

The obtained relative partial molar functions of zinc in the alloys are presented in Table 2.

According to the phase diagram of the Zn-Sb binary system [22], ZnSb compound is the richest in the antimony phase in the system and is in ZnSb + Sb equilibrium with it. Hence the partial molar functions of zinc in this region are thermodynamic functions of the following virtual-cell reaction:

$$Zn(s) + Sb(s) \rightarrow ZnSb(s)$$
. (6)

This reaction is similar to the reaction of the formation of ZnSb compound from its elemental components. Therefore, the corresponding partial molar functions of zinc in the ZnSb + Sb phase area are standard thermodynamic functions of the formation of ZnSb.

The partial molar functions of zinc in the Zn_4Sb_3 + ZnSb two-phase field (Table 2) are thermodynamic functions of the following virtual-cell reaction:

$$Zn(s) + 3ZnSb(s) \rightarrow Zn_{A}Sb_{z}(s).$$
 (7)

In accordance with the reaction (7), the following equations were used to calculate the standard Gibbs free energy, enthalpy, and entropy of the formation of the Zn_4Sb_3 intermediate compound:

$$\Delta_f Z^0(\mathrm{Zn_4Sb_3}) = \Delta \overline{Z}_\mathrm{Zn} + 3\Delta_f Z^0(\mathrm{ZnSb}), \qquad (8)$$
 where $Z \cong G$, H , or S .

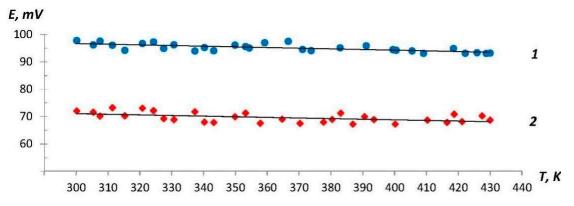


Fig. 2. $E \sim f(T)$ plots for the samples along the system Zn-Sb: (a) ZnSb +Sb phase area; (b) ZnSb + Zn₄Sb₃ phase area

Table 1. Temperature dependencies of the *emf* for the cells of the type (1) in some phase areas of the Zn–Sb system at the 300-430 K temperature interval

Phase region	$E, mV = a + bT \pm 2S_{E}(T)$
ZnSb(s) + Sb(s)	$103.95 - 0.0243T \pm 2 \left[\frac{1.04}{30} + 2.1 \cdot 10^{-5} (T - 363.5)^{2} \right]^{1/2}$
$Zn_4Sb_3(s) + ZnSb(s)$	$77.78 + 0.0225T \pm 2 \left[\frac{2.26}{30} + 4.7 \cdot 10^{-5} (T - 363.2)^{2} \right]^{1/2}$

Table 2. Partial molar thermodynamic functions of zinc in the alloys of the Zn-Sb system at 298 K

Phase region	$-\Delta ar{G}_{\!\scriptscriptstyle \mathrm{Zn}}$, kJ/mol	$\Delta ar{H}_{ m Zn}$, kJ/mol	$\Delta \overline{S}_{\rm Zn}$, J/(mol·K)
ZnSb(s) + Sb(s)	18.66±0.14	20.06±0.64	-4.69±1.76
$Zn_4Sb_3(s) + ZnSb(s)$	13.71±0.20	15.01±0.98	-4.36±2.66

Absolute entropies of the ZnSb and Zn₄Sb₃ compounds have been calculated using the following equations:

$$S^{0}(ZnSb) = \Delta \overline{S}_{Zn} + S_{Zn}^{0} + S_{Sb}^{0}$$
 (9)

$$S^{0}(Zn_{4}Sb_{3}) = \overline{\Delta S}_{Zn} + S^{0}_{Zn} + 3S^{0}_{ZnSh}.$$
 (10)

Errors were calculated by the error accumulation method. Absolute entropies of the elementary zinc and antimony used for calculations were taken from [25]: $S^{\circ}(Zn) = 41.63\pm0.13 \text{ J/(mole\cdot K)}$; $S^{\circ}(Sb) = 45.69\pm0.63 \text{ J/(mole\cdot K)}$. Calculated

standard integral thermodynamic functions of both compounds along with available literature data are tabulated in Table 3.

The $\Delta_f G^\circ$ values in the table related to the studies with the *emf* measurements [38-40, 46] were calculated by us for 298 K based on the results of those studies. Comparative analysis of the values in the table shows that the results of the current study for the ZnSb compound are in good agreement with the results of the *emf* measurements given in [37, 38], obtained by vapor pressure measurement method [41], as well as

Table 3. Standard integral thermodynamic functions of zinc antimonides

Compound	$-\Delta_{_f}G^{\circ}$	−Δ _f H° kJ/mol	$\Delta_f S^{\circ}$ J/(mol·K)	S°	Reference, method
ZnSb	18.7±0.2	20.1±0.7	-4.7±1.8	82.6±2.5	This work, emf
	19.89	21.80			[37], emf
	17.75	19.10			[38], <i>emf</i>
	16.28	16.46			[39], <i>emf</i>
	16.53	17.15			[40], <i>emf</i>
	17.68	19.00			[41], vap. press.
		12.8			[42], calorimetry
		74.47			[43], calorimetry
	19.5	22.49			[17], optimization
	20.36	22.98			[44], optimization
		22.92			[45], optimization
		19.0±1.3		82.6±1.7	[24], recommend
	17.40	16.74		89.54±2.1	[25], recommend
	17.72	18.93		83.08	[26], recommend
Zn_4Sb_3	69.7±0.7	75.2±3.0	-18.5±7.9	285.1±10.4	This work, emf
	65.30	74.90			[37], <i>emf</i>
	55.48	37.24			[39], <i>emf</i>
	56.82	54.79			[46], emf
	61.31	65.57			[44], optimization
	65.02	65.65			[47], optimization

with the ones thermodynamically optimized in [17, 44, 45]. The results obtained from other *emf* studies are somewhat lower. Quantities obtained during two different calorimetric studies differ from each other by more than 5 times, and hence they are naturally insignificant. It should also be noted that for the ZnSb compound, the quantities provided in the reference books [24-26] are slightly different than our results. Among them, the results of [24] and [26] are in better agreement with the current study.

Thermodynamic properties of the $\rm Zn_4Sb_3$ compound were experimentally studied only using the *emf* method [37, 39, 46] until our studies and their thermodynamic optimization were carried out by [44, 47]. As can be seen from Table 3, our results are closer to [37] and also agree with the optimized thermodynamic quantities.

Thus, the obtained standard thermodynamic functions have high accuracy and minimize existing contradictions, especially for the standard entropy and Gibbs free energy of formation for both compounds.

4. Conclusion

The present contribution shares the results of the thermodynamic study of the ZnSb and Zn₄Sb₇ compounds by a low-temperature emf method. Using samples from the ZnSb+Sb and Zn₄Sb₇+ZnSb phase areas of the system as a right electrode, the linear equations of temperature dependences of the emf have been obtained. From these linear equations, the partial molar thermodynamic functions of zinc in alloys have been calculated. Based on the T-x diagram of the Zn-Sb system, the potential generating reactions for both binary compounds have been defined and consequently, their standard integral thermodynamic functions have been determined. The set of obtained standard thermodynamic quantities is of high accuracy and a new contribution to the thermodynamics of zinc stibnites. These results allow us to minimize the contradictions between the existing literature data.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Refinement of the phase diagram of the MnSe-In₂Se₃ system and the crystal structures of MnIn₂Se₄ and Mn₂In₂Se₅ compounds

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Abstract

Complex chalcogenides based on transition elements, in particular ternary compounds of the AB_2X_4 type (M = Mn, Fe, Co, Ni; B = Ga, In, Sb, Bi; X = S, Se, Te) are among the important functional materials. Compounds of this class exhibit the phenomena of electronically or optically controlled magnetism and are very promising for the creation of lasers, light modulators, photodetectors, and other functional devices controlled by a magnetic field. Recent studies demonstrated that these compounds can also find application in photocatalysis, photovoltaics, and thermoelectric converters.

The study presents new data on phase equilibria in the MnSe– ${\rm In_2Se_3}$ system, obtained by differential thermal analysis, X-ray phase analysis, and scanning electron microscopy. Two ternary compounds, MnIn₂Se₄ with congruent melting at 1193 K and Mn₂In₂Se₅, melting incongruently at 1196 K, were formed in the system. The first is a phase of variable composition and has a 5–6 mol. % homogeneity region towards an excess of ${\rm In_2Se_3}$. Based on powder diffraction data, the Rietveld method was used to refine the crystal structures and lattice parameters of both ternary compounds.

Keywords: Manganese-indium selenides, Phase equilibria, Homogeneity region, Crystal structure, Rietveld method *Funding:* This work is supported by the Azerbaijan Science Foundation – Grant No AEF-MCG-2022-1(42)-12/10/4-M-10.

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1. Introduction

Complex transition metal chalcogenides, in particular compounds of the AB₂X₄ and A₂B₂X₅ types (A = Fe, Mn, Co; B = Al, Ga, In, Sb, Bi; X = S, Se, Te) with a layered structure are among the important materials possessing thermoelectric [1, 2], magnetic [3–9], optical, and other functional properties [10-13]. This makes them very promising for use in the creation of lasers, light modulators, photodetectors, thermoelectric, and other functional devices. In addition, they are objects of intensive research as magnetic topological insulators, combining the properties of an antiferromagnet and a topological insulator, and are extremely promising for use in spintronics, quantum computing, and information processing devices [14-22].

A number of recent studies have shown that some compounds of the AB₂X₄ type and heterojunctions based on them can find application in photocatalysis and photovoltaics, in particular, in photocatalytic water splitting [23–27]. According to the results of [28–30], some compounds of the above type with a spinel structure are good candidates for use as a new type of anode materials for the stable storage of ions in lithium (sodium) ion batteries.

The results described above show the relevance of research aimed at obtaining and studying the properties of new complex layered chalcogenides of transition elements. Modification of such compounds by alloying and obtaining solid solutions can be used to optimize their functional properties [31–36].

The development of methods for the targeted synthesis of complex chalcogenide compounds and phases of variable composition is based on data of phase equilibria in the corresponding systems. Phase diagrams provide valuable information on the nature of formation, thermal stability, phase transformations, primary crystallization regions, and the homogeneity of intermediate phases [37–43].

Previously, studies of phase equilibria in a number of quasi-ternary $MX-Ga_2X_3-In_2X_3$ systems (M=Mn,Fe;X=S,Se,Te) were performed, in search of the physicochemical foundations for creating new magnetic semiconductors [44–50]. In the indicated systems, new phases of variable composition based on ternary AB_2X_4 compounds have been identified, their primary crystallization fields and homogeneity regions were determined.

Preliminary experimental results in the study of the MnSe– Ga_2Se_3 – In_2Se_3 system, revealed their discrepancy with the known [51] phase diagram of the MnSe– In_2Se_3 boundary system. Taking this into account these findings, in this study we have undertaken a repeated study of phase equilibria in the MnSe– In_2Se_3 system.

The initial compounds of the studied system were investigated in detail. The MnSe compound melts congruently at 1875 K and has three modifications: a stable low-temperature α -MnSe crystallizes in a cubic NaCl-type structure. The β -MnSe and γ -MnSe phases are unstable. The first phase crystallizes in a cubic structure of the sphalerite type, and the second crystallizes in a hexagonal structure of the wurtzite type [52–55], (Table 1).

The In_2Se_3 compound melts with an open maximum at 1158 K and undergoes three polymorphic transformations (473, 920, and 1023 K) [52, 53]. The types and parameters of crystal lattices of all four crystalline modifications of In_2Se_3 are described in detail in [56], (Table 1).

Table. 1. Crystallographic data of MnSe and In₂Se₃

Соединение	Type and parameters of the crystal lattice, nm	Ref.
MnSe-rt	Kubik, Sp.Gr. $Fm\bar{3}m$; $a = 0.5456$	[54]
MnSe-ht1	Kubik, Sp.Gr. $F\bar{4}3m$; $a = 0.583$	[55]
MnSe-ht2	Heksaqonal, Sp.Gr. $P63mc$; $a = 0.413$, $c = 0.673$	[55]
In ₂ Se ₃ -rt	Heksaqonal, Sp.Gr. $R3m$; $a = 0.405$, $c = 2.877$	[56]
In ₂ Se ₃ -ht1	Heksaqonal, Sp.Gr. $P6_{\varsigma}$; $a = 0.711$, $c = 1.930$	[56]
In ₂ Se ₃ -ht2	Heksaqonal, Sp.Gr. $P6_1$; $a = 0.7133$, $c = 1.934$	[56]
In_2Se_3 - ht3	Heksaqonal, Sp.Gr. $P6_1$; $a = 0.4014$, $c = 0.964$	[56]

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2. Experimental

2.1. Synthesis

The MnSe and In₂Se₃ compounds were synthesized by the direct interaction of stoichiometric quantities of high-purity elemental components (manganese – 99.95%, indium – 99.99% and selenium – 99.99% from Alfa Aesar) in sealed quartz glass ampoules evacuated to ~ 10^{-2} Pa. The syntheses were carried out in a two-zone inclined furnace. The lower "hot" zone was heated to 1200 K, and the upper "cold" zone was heated to 900 K, which is slightly lower than the boiling point (958 K) of elemental selenium [57]. In order to avoid the interaction of quartz with manganese, MnSe synthesis was carried out in a graphitized ampoule.

The individuality of the synthesized compounds was controlled by differential thermal analysis (DTA) and X-ray diffraction (XRD) methods. Temperatures of polymorphic transitions and melting of In_2Se_3 , determined based on the DTA heating curves, coincided with the available literature data [52, 53]. The following crystallographic parameters were obtained by interpretation of the powder X-ray diffraction patterns: MnSe – cubic, sp. gr. $Fm\bar{3}m$, a = 0.54542(4) nm, RT- In_2Se_3 – hexagonal, sp.gr. $R\bar{3}m$, a = 0.40804(5), c = 2.8712(14) nm, which are in good agreement with the literature data [54–56], (Table 1).

Alloys of the MnSe– In₂Se₃ system were prepared by fusing the initial binary compounds in various ratios in evacuated quartz ampoules at 1200–1300 K, followed by thermal annealing at 800 K for 500 hours and cooling in a switched-off furnace.

2.2. Research methods

The studies were carried out using DTA, X-ray diffraction and scanning electron microscopy (SEM) methods. The DTA of samples weighing $0.1-0.3\,\mathrm{g}$ in evacuated quartz ampoules was carried out using a Netzsch STA 449 F3 unit (platinum – platinum/rhodium thermocouples) in the temperature range from room temperature to ~ 1450 K with a heating rate of $10~\mathrm{K}\cdot\mathrm{min}^{-1}$. The temperature measurement accuracy was within $\pm 2~\mathrm{K}$.

The X-ray powder diffraction analysis of powder samples was performed using a D2 Phaser diffractometer (Bruker, Germany; $CuK\alpha$ -radiation, angle range $5^{\circ} \le 20 \le 80^{\circ}$, scanning rate of

0.03°×0.2 min). The crystal lattice parameters of ternary compounds were refined by the Rietveld method using the EVA and TOPAS 4.2 programs. The SEM analysis was performed using the Tescan Vega 3 SBH scanning electron microscope.

3. Results and discussion

The joint processing of DTA, X-ray diffraction, and SEM data for synthesized and annealed alloys allowed us to obtain a new, more precise pattern of phase equilibria in the MnSe – In_2Se_3 system. Below in the text, in the figures and in the tables the following phase designations are used: α and γ -solid solutions based on the MnSe and MnIn $_2Se_4$; β_1 , β_2 , β_3 and β_4 -solid solutions based on high-temperature, two intermediate and low-temperature modifications of the In_2Se_3 compound, respectively.

The X-ray diffraction method showed that alloys of composition 33.3 ($Mn_2In_2Se_5$) and 50 mol. % In_2Se_3 ($MnIn_2Se_4$) have individual diffraction patterns that differ from those of the original compounds. Diffraction patterns of other intermediate alloys consisted of two-phase mixtures $\alpha + Mn_2In_2Se_5$, $Mn_2In_2Se_5 + MnIn_2Se_4$ (γ) and $\gamma + \beta_4$ respectively (Fig. 1).

Based on powder diffraction patterns, the Rietveld method was used to refine the crystal structures of both ternary compounds. The experimental and calculated diffraction patterns, as well as the differences in their intensities are shown in Fig. 2 and 3. The refined parameters of the elementary cells are shown in Table 2, and the atomic positional parameters are shown in Tables 3 and 4. The crystal structures of the MnIn₂Se₄ and Mn₂In₂Se₅ compounds are shown in Fig. 4. Both compounds are layered phases of the van der Waals type. Their structural blocks are fragments consisting of 7 and 9 atomic layers, in which the atoms alternate as Se-In-Se-Mn-Se-In-Se and Se-In-Se-Mn-Se-In-Se respectively. In both structures, Mn atoms are located at the centers of octahedra, and In atoms are located at the centers of tetrahedra formed by Se atoms. These blocks are bound together by van der Waals bonds. The results obtained for polycrystalline samples using the Rietveld method are in good agreement with the data of studies [58, 59], obtained based on diffraction data of single crystals.

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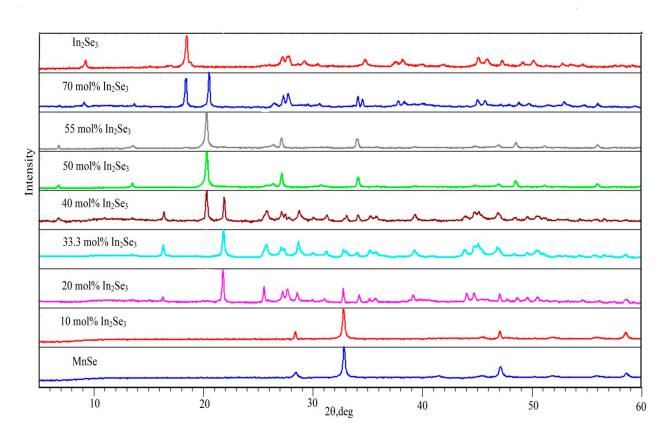
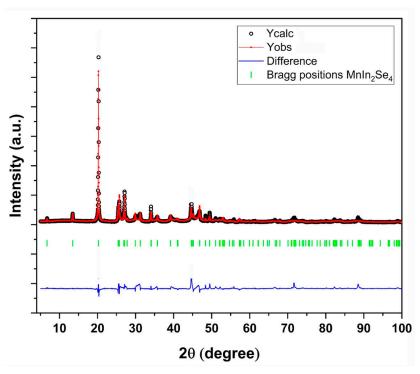


Fig. 1. The XRD patterns of alloys of the MnSe-In₂Se₃ system



 $\textbf{Fig. 2.} \ \textbf{Experimental and calculated diffraction lines of MnIn}_2 \textbf{Se}_4, \text{ as well as the differences in their intensities}$

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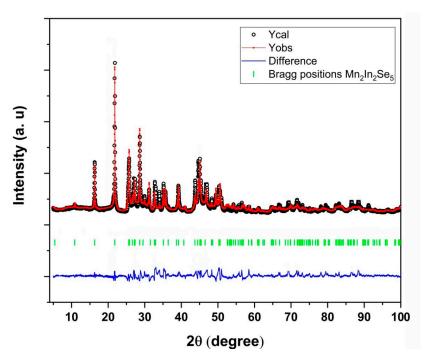


Fig. 3. Experimental and calculated diffraction lines of $Mn_2In_2Se_5$, as well as the differences in their intensities

Table. 2. Refined parameters of the structure of MnIn₂Se₄ and Mn₂In₂Se₅

Structure parameters	MnIn ₂ Se ₄	Mn ₂ In ₂ Se ₅
Space group	$R\bar{3}m$	$R\bar{3}m$
Cell parameters:		
a (nm)	0.405289(45)	0. 402240(57)
<i>c</i> (nm)	3.94594(44)	4.87486(79)
The cell volume (nm³)	0.56243(17)	0.68307(22)
Density (g/cm³)	5.318(91)	5.36014(50)
R-Bragg (%)	0.873	0.428

Table. 3. Atomic positional parameters in MnIn₂Se₄

Atoms	Multiplicity of positions	X	у	z	Atom type	Relative occupation
In1	6	0.00000	0.00000	0.76964(31)	In ⁺³	1.000(42)
Se2	6	0.00000	0.00000	0.70467(46)	Se	1
Se1	6	0.00000	0.00000	0.12160(56)	Se	1
Mn	3	0.00000	0.00000	0.00000	Mn ⁺²	1.001(68)

Table.4. Atomic positional parameters in Mn₂In₂Se₅

Atoms	Multiplicity of positions	X	у	z	Atom type	Relative occupation
Se2	6	0.00000	0.00000	0.39354(22)	Se	1
Mn1	6	0.00000	0.00000	0.70092(31)	Mn ⁺²	1
Se1	6	0.00000	0.00000	0.13546(20)	Se	1
In2	6	0.00000	0.00000	0.55513(13)	In ⁺³	1
Se3	3	0.00000	0.00000	0.00000	Se	1

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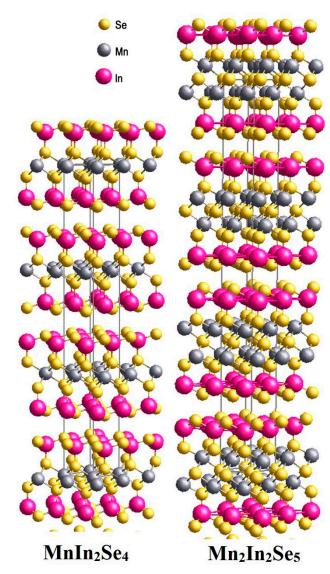


Fig. 4. Three-dimensional view of the structures of MnIn₂Se₄ and Mn₂In₂Se₅

It should be noted that the $MgIn_2Se_4$ and $ZnIn_2S_4$ compounds are also characterized by a similar structure, however, the authors [60] chose an asymmetric space group $R\bar{3}m$ to describe the structure of $MgIn_2Se_4$.

Interpretation of DTA data on the heating of annealed alloys of MnSe– ${\rm In_2Se_3}$ system (Table 5), taking into account the above presented XRD results, allowed us to construct a T-X phase diagram (Fig. 5). According to our data, the MnSe– ${\rm In_2Se_3}$ system can be considered quasi-binary, since the compositions of all phases in equilibrium are on its T-X planes. The system is characterized by the formation of two intermediate compounds: the ${\rm MnIn_2Se_4}$ compound melts congruently at 1195 K and has a wide homogeneity region

Table. 5. DTA results of alloys of the MnSe–In₂Se₃ system

Composition, mol. % In ₂ Se ₃	Thermal effects, K
10	-
20	1195
30	1197; 1197–1440
33.3	1196; 1196–1390
35	1188–1196; 1196–1360
40	1188 –1196; 1196-1285
45	1188
50	1195
55	1180-1193
60	1150-1190
70	462; 1148; 1148–1185
80	460; 1148; 1148–1176
90	463; 910; 1005; 1150
95	465; 910; 1005;1148–1155
100	475; 920; 1023; 1163

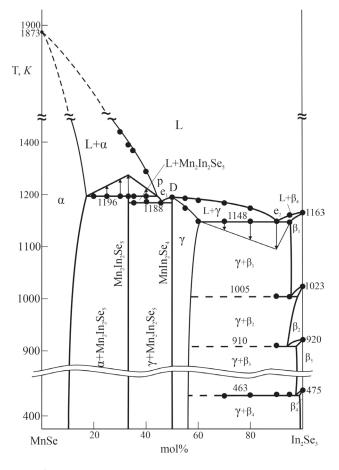


Fig. 5. Phase diagram of the MnSe-In₂Se₂ system

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(γ-phase) towards an excess of the In_2Se_3 . The second compound of the composition $Mn_2In_2Se_5$ melts with decomposition according to the peritectic reaction $L + a \leftrightarrow Mn_2In_2Se_5$ at 1196 K. The peritectic point has a composition of 42 mol. % In_2Se_3 , the γ-phase is in eutectic equilibrium with the neighboring phases:

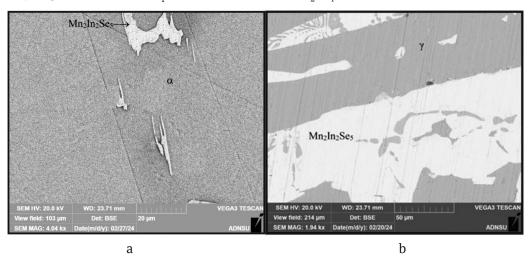
$$\begin{split} L &\leftrightarrow Mn_2In_2Se_5 + \gamma \\ &(\text{point } e_1, 45 \text{ mol. } \% \text{ } In_2Se_3, 1188 \text{ } K) \\ L &\leftrightarrow \beta_1 + \gamma \text{ } (\text{point } e_2, 90 \text{ mol. } \% \text{ } In_2Se_3, 1148 \text{ } K). \end{split}$$

The compositions of the peritectic and eutectic points, as well as the homogeneity regions of the phases, were refined by the plotting of the Tamman's triangle (Fig. 5). It has been established that the homogeneity region of the α -phase at the peritectic temperature is 15 mol. %, and at eutectic temperature e_2 homogeneity regions of γ - and β_1 -phases reach

10 and 5 mol. %, respectively. With a decreasing temperature of the homogeneity region the α -and γ -phases were somewhat narrowed and, according to X-ray diffraction data (Fig. 1), at room temperature, they were ~10 and 5 mol. %.

The phase compositions of the alloys, in particular the areas of phase homogeneity, were confirmed by SEM results (Fig. 6). As can be seen, the SEM images are in accordance with the phase diagram: the alloy with the composition $55 \, \text{mol.} \%$ In_2Se_3 , as well as the alloy with the stoichiometric composition MnIn_2Se_4 are single-phase, and the other three samples from different two-phase regions of the phase diagram are two-phase.

The obtained pattern of phase equilibria in the MnSe–In₂Se₃ system differed significantly from the data obtained in [51]. The phase diagram presented in [51] reflected only one ternary compound: MnIn₂Se₄ with congruent melting at ~1200 K. In



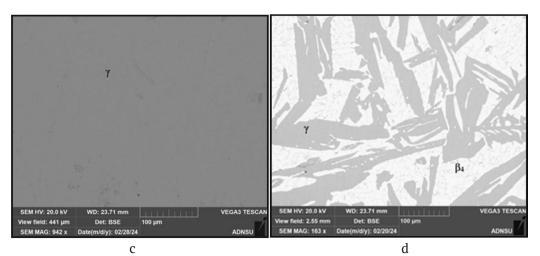


Fig. 6. SEM images of some single- and two-phase alloys of the MnSe-In₂Se₃ system: a) 15; b) 40; c) 55 and d) 70 mol. % In₂Se₃

addition, according to data [51], solubility based on MnSe was practically absent, and homogeneity regions based on various modifications of $\rm In_2Se_3$ ranged from 5 (low temperature) to 25 mol. % (high temperature). Our data also differ significantly from data of [51] for the coordinates of eutectic and eutectoid equilibria.

4. Conclusions

Thus, we present a new refined version of the phase diagram of the MnSe-In $_2$ Se $_3$ system, on which, in contrast to the data [51], in addition to MnIn $_2$ Se $_4$, a ternary compound of the Mn $_2$ In $_2$ Se $_5$ composition, melting with decomposition according to a peritectic reaction at 1196 K, was also formed. According to our data, MnIn $_2$ Se $_4$ melts congruently at 1193 K and is a phase of variable composition, the homogeneity region at room temperature is ~5 mol. %. The crystal structures and lattice parameters of the MnIn $_2$ Se $_4$ and Mn $_2$ In $_2$ Se $_5$ compounds were refined based on the XRD data, using the Rietveld method. Both ternary compounds were shown to have a layered structure and crystallize in the $R\bar{3}m$ space group.

Author contributions

F. M. Mammadov – research concept, discussion of results, writing the article; S. Z. Imamaliyeva – processing of the obtained results and participation in writing the article; E. N. Ismailova – literature search and participation in experiments; I. R. Amiraslanov – conducting structural studies; E. I. Akhmedov – processing the obtained results and writing the article; M. B. Babanly – scientific leadership, scientific editing.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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A universal algorithm for the calculation of vapor-liquid equilibrium diagrams in quasi-simple multicomponent systems

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Abstract

The purpose of the study was to consider isothermal vapor-liquid diagrams of quasi-simple systems and to develop a universal algorithm for the calculation of isothermal vapor-liquid diagrams of these systems independent of the type of valence of the electrolyte, the number of components in the system, and the types of solid solutions. The suggested analogues of the three Gibbs–Konovalov and Gibbs–Roozeboom laws are true when moving along the univariant equilibrium lines on the solubility diagrams of systems with a random number of components.

The study did not involve any experiments. The suggested algorithm was applied for the description of solubility (solid-liquid) diagrams and vapor-liquid equilibrium diagrams of three- and four-component systems with one, two, or three volatile components. In all the cases, the results of thermodynamic first-principles calculations agreed well with the experimental data presented in the literature.

Both the experimental data presented in the literature and the results of the thermodynamic first-principles calculation performed by the authors are also in good agreement with the suggested analogues of the Gibbs–Konovalov and Gibbs–Roozeboom laws.

Keywords: Zdanovskii's rule, Quasi-simple systems, Vapor-liquid equilibrium diagrams, Partial molar Gibbs energy, Analogues, Gibbs–Konovalov's laws, Gibbs–Roozeboom rules

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1. Introduction

1.1. Main thermodynamic properties of quasi-simple systems

We shall start by introducing the main definitions. Water-salt systems with three or more components (or, in the general case, solvent-solute systems) comply with the socalled Zdanovskii's rule, if isoactivates of the solvent (hereinafter, water – W) are line segments (n = 3), plane sections (n = 4), or hyperplane sections $(n \ge 5)$ [1–6]. Section 1.1 of the article is largely based on the terms used in [6]. The article considers isothermal-isobaric conditions (T, P = const). Water isoactivates $(a_{w} = \text{const})$ are isopotentials ($\mu_w = \text{const}$; μ_w is the chemical potential), isobars of partial pressures of water $(P_{w} = \text{const})$, and isotherm-isobars of a two-phase vapor-liquid equilibrium (l-v) in the region of homogeneous liquid solutions at the same time [1-6]. The equation of the isoactive line divided into segments for a ternary system complying to Zdanovskii's rule is mathematically expressed by the following relation [1, 3, 4, 6]:

$$m_1 / m_1^0 + m_2 / m_2^0 = 1$$
, (1)

where: $m_i \bowtie m_i^0$ is the molality of the i-th component of a ternary and a binary system with the same value of μ_{w} , respectively. When dealing with a random n-component system, the equation of the isoactive hyperplane is presented as follows:

$$\sum_{i=1, i \neq W}^{n-1} m_i / m_i^0 = 1 ; (2)$$

or

$$\sum_{i=1, i \neq W}^{n-1} Y_i = 1; \ Y_i = m_i / m_i^0.$$
 (3)

It is important that the isopotentials of other (i.e. salt) components are not line segments or hyperplane sections. In particular, they are crystallization curves of these anhydrous components on the solubility diagrams of ternary systems and cannot be linear by nature.

Hereinafter, in order to avoid ambiguity, we will refer to systems as being **quasi-simple** if

the **isoactivates** of at least one component are **linear**. In this case, the properties of systems with three or more components are determined by the properties of their binary subsystems. This is accounted for by the Gibbs-Duhem equation and the conditions of cross product derivation for the liquid phase under isothermal–isobaric conditions [4, 7-9]:

$$SdT - VdP + \sum_{i=1}^{n} n_i d\mu_i = 0$$
(4)

or when T, P = const:

$$\sum_{i=1}^{n} n_i d\mu_i = 0, (5)$$

$$\frac{\partial^2 G}{\partial n_i \partial n_j} = \frac{\partial^2 G}{\partial n_i \partial n_j} \tag{6}$$

or when T, P = const:

$$\partial \mu_i / \partial n_i = \partial \mu_i / \partial n_i$$
. (7)

Thus, in particular, excess thermodynamic functions of multi-component liquid phases (activity of the components – a_i , activity coefficients – γ_i , water osmotic coefficients – ϕ), as well as equilibrium diagrams (l-v) and (s-l) of multi-component systems should be calculated based on the data regarding the thermodynamic functions of their binary subsystems. The main question can be formulated as follows: "Which unique properties of thermodynamic potentials of a phase of variable composition, i.e. a liquid (in our case, they are the Gibbs energy – G or partial potentials - Korzhinsky's potentials [4, 10] – $G^{[W]}$), make a system quasi-simple?". This question can be answered by determining the type of functional dependence of excess thermodynamic functions on the composition of multi-component systems.

Let's briefly review the main points presented in [6]. Let's consider an n-component system: 1-2-3...n-1-W (W is a solvent (water), components 1, 2, 3... n-1 are dissolved electrolytes (salts)) with T, P = const. Let's introduce the partial Gibbs energy [4, 10–12]:

$$G^{[W]} = G - n_W \mu_W = \sum_{i=1}^{n-1} n_i \mu_i,$$
 (8)

where $G^{[W]}$ is the partial Gibbs energy, μ_i is the chemical potential of the i-th solute; μ_w is the chemical potential of water. $G^{[W]}$ is characteristic for the set of variables $(T, P, n_i \ (i \le n-1), \mu_W)$. Let's now consider the quasi-average partial molar Gibbs energy calculated for 1 mole of solutes – $G^{(W)}$:

$$G^{(W)} = \sum_{i=1, i \neq W}^{n-1} Z_i \mu_i,$$
 (9)

where: Z_i is the mole fraction of the i-th solute in the concentration space without solvents:

$$Z_{j} = \frac{n_{j}}{\sum_{i=1, i \neq W}^{n-1} n_{i}}; \sum_{i=1, i \neq W}^{n-1} Z_{i} = 1.$$

When T,P = const, $G^{(W)}$ is characteristic for the set of variables $(Z_1,Z_2,...,Z_{n-2},\mu_W)$. Let's introduce the function $\Delta G^{(W)}_{mix}$, which

Let's introduce the function $\Delta G_{mix}^{(W)}$, which reflects changes in the average partial molar Gibbs energy when mixing (n-1) binary solutions (i-W), each of which contains Z_i moles of the i-th component [6]. Let all the mixed binary solutions have the same chemical potential of the solvent (i.e. are isopiestic), $\mu_W^{(1-W)} = \mu_W^{(2-W)} = \dots = \mu_W^{(n-1-W)} = const$. Let's assume that after mixing, the chemical potential of the solvent in the resulting multi-component system remains the same, $\mu_W^{(mult)} = const$ i.e. the system complies with Zdanovskii's rule. Then function $\Delta G_{mix}^{(W)}$ no longer depends on μ_W but rather on the variables of the composition $-Z_i$: $\Delta G_{mix}^{(W)}(Z_1, Z_2 \dots Z_{n-2})$ [6]. Obviously, in the general case:

$$\frac{\Delta G_{mix}^{(W)}}{RT} = \sum_{i=1}^{n-1} Z_i \left(\ln a_i^{(mult)} - \ln a_i^0 \right), \tag{11}$$

where: $a_i^{(mult)}$ is the activity of the i-th solute after mixing; a_i^0 is the activity of the i-th solute in the binary system (i-W) with the same value of μ_w before mixing [6].

Let's assume that in quasi-simple systems, the function $\Delta G_{mix}^{(W)}$, corresponding to the mixing of binary solutions (*i-W*) with the same μ_W , provided that μ_W remains unchanged

after mixing, is only determined by the entropy of ideal mixing of binary solutions or [6]:

$$\frac{\Delta G_{mix}^{(W)}}{RT} = \sum_{i=1}^{n-1} Z_i \ln Z_i.$$
 (12.1)

In other words, the excess partial Gibbs free energy of the mixed isopiestic binary solutions ($\Delta G_{mix}^{(W)-ex}$) should be zero:

$$\frac{\Delta G_{mix}^{(W)-ex}}{RT} = 0. \tag{12.2}$$

By comparing equations (11) and (12.1) we obtain:

$$\ln a_i^0 + \ln Z_i = \ln a_i^{(mult)}$$
 (при $\mu_W = const$), (12.3)

$$\ln m_i = \ln Z_i + \ln m_i^0, \tag{13}$$

where: m_i^0 is the molality of the i-th component in the binary solution (before mixing) with constant μ_w ; m_i is the molality of the i-th component in the multi-component solution (after mixing) with the same value of μ_w .

In [6], the main equations (12.1) - (12.3) are substantiated in a different way. Let's accept the classical definition of the number of components in a system (n) as a number of ion and molecular forms in the system, whose masses can change independently [4, 6, 9]. Let's assume that the studied system is quasi-simple. Let the chemical potential of the solvent be $\mu_w = const$. Then binary solutions (i-W) with identical μ_w can be considered to be **new individual components.** If, with the same μ_w , these components are mixed as components of an ideal solution, then equations (12.1)-(12.3) are true.

[6] presents the results of the first-principles calculation of equations (12, 13) based on the data regarding the corresponding binary subsystems of the solubility diagrams of three- and four-component systems with crystallization of anhydrous salt components, their crystalline hydrates, ternary compounds of constant compositions, and solid solutions with isovalent substitutions. In all the cases, the calculations agreed well with the experimental data presented in the literature.

1.2. Calculation of vapor-liquid equilibria of quasi-simple systems

The article focuses on first-principles calculation of vapor-liquid equilibrium diagrams of three- and more component systems with volatile components based on equations (12, 13) and the data regarding their binary subsystems.

Let the chemical potential or activity of water be constant ($\mu_w = const$; $\ln a_w = const$).

According to Henry's law, for the partial vapor pressure of a volatile component of a multicomponent system (hereinafter, assuming that the vapor phase is ideal) the following is true [7, 9, 13]:

$$\ln P_{i} = \ln a_{i}^{(mult)} + \ln K_{H i}^{m}, \qquad (14.1)$$

where: P_i is the partial pressure of the i-th volatile component (mm Hg), $a_i^{(mult)}$ is the activity of the i-th component on the molality scale, and $K_{H,i}^m$ is the Henry's law constant of the i-th component on the molality scale. Or, taking into account (12.3):

$$\ln P_i = \ln a_i^0 + \ln Z_i + \ln K_{H,i}^m$$
 (при $\mu_W = const$), (14.2) where: a_i^0 is the activity of the *i*-th solute in the binary system (*i*–*W*) with the same μ_w before mixing [6].

The partial pressure of the solvent P_{w} is determined according to Raoult's law [7, 9, 13]:

$$\ln P_W = \ln a_W + \ln P_W^{(0)}, \tag{14.3}$$

where $P_W^{(0)}$ is the vapor pressure above the pure solvent under specific conditions (mm Hg). For instance, with water being the solvent, at 25 °C $P_W^{(0)} = 23.76$ mm Hg.

Thus, according to Dalton's law, the total vapor pressure above the solution is P_{sum} , [7, 9, 13]:

$$P_{sum} = \sum_{i=vc} a_i^0 Z_i K_{H,i}^m + a_W P_W^{(0)}, \qquad (14.4)$$

where all the volatile components except for the solvent are summarized. The composition of the solution (m_i) is set by equation (13), based on the data regarding binary isopiestic concentrations (m_i) .

2. Binary systems HCl – H_2O and HBr – H_2O at 25 °C

This section, though not entirely within the scope of the study, is still important, because it demonstrates the accuracy of the calculation of partial pressures of the components performed by means of a classical model of electrolyte solutions used to describe binary systems, namely the Pitzer model [14, 15] with three and four parameters. As a result, we selected two binary systems with two volatile components: a solvent ($W = H_2O$) and a volatile acid, HCl – H_2O , HBr – H_2O at 25 °C. The data regarding the binary Pitzer parameters is presented in Table 1. We should note that the said parameters quite accurately describe the concentration dependence of the water osmotic coefficients on the molality of the binary solutions $\varphi(m)$ almost over the whole concentration range $m = 0 \div 16 \text{ mol/kg H}_{2}O$, including the region of azeotropes (Az), where the vapor pressure is low $(P_T^{Az} = \min)$. The corresponding Henry constants $(K_{H,HCI}^m; K_{H,HBr}^m)$ are given in Table 2. Fig. 1 and 2 present the results of calculations and the experimental data regarding the partial pressures of halogen hydracids and water and the total vapor pressure above the solutions studied in [16, 17]. The figures show that the experimental and calculated data are in good agreement, including in the region of binary azeotropes – Az_i .

Table 1. Binary Pitzer parameters at 25 °C

Cystom	Binary parameters (r.u.)							
System	$\beta^{(0)}$	$\beta^{(1)}$	$\beta^{(2)}$	C^{φ}	$\alpha_{_1}$	α_2		
NaCl-H ₂ O	0.0765	0.2664	0	0.00124	2.0	_		
KCl-H ₂ O	0.0484	0.2122	0	-0.00084	2.0	_		
LiCl-H ₂ O	0.27909	1.5093	-0.78110	-0.00765	2.0	1.0		
LiNO ₃ -H ₂ O	0.09642	0.03872	0.02704	-0.01010	2.0	-0.4		
HCl-H ₂ O	0.1769	0.2972	0	0.00072	2.0	_		
HBr-H ₂ O	0.2259	0.1372	0.0289	-0.00167	2.0	1.0		

Table 2. Henry constants (on molality scale) for systems HCl – H₂O, HBr – H₂O at 25 °C

System	HCl-H ₂ O	HBr-H ₂ O	
$K_H^m (mm Hg)$	1.99·10 ⁻⁴	4.04.10-6	

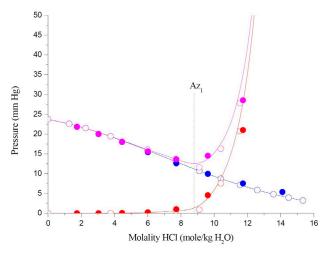


Fig. 1. The data calculated based on the Pitzer equations (lines and open circles) and the experimental data [16] (solid circles) regarding the concentration dependence of the pressures: H_2O (blue), HCl (red), and total (violet). Az_1 – azeotrope

3. Ternary systems with one volatile component – the solvent (for instance, the LiCl – LiNO $_3$ system) at 25 °C

This section considers a system with one volatile component, namely the LiCl – LiNO, system at 25 °C. The system is quite trivial. Its equilibrium diagram is represented by a set of linear isoactivates of water. Therefore, we assume that the system complies with Zdanovskii's rule over the whole concentration range. The calculated and experimental isoactivates of water in the studied system are presented in Table 3 and Fig. 3. The experimental data regarding the water activity was obtained by means of the isopiestic method using a variation of Resnik's measurement unit [18]. Binary solutions of the LiCl – H₂O system were used as reference solutions. The solubility was calculated based on the data regarding the binary subsystems using an algorithm of calculation of solubility diagrams suggested in [6]. Fig. 3 shows that the calculated data on the isoactivates of water agrees well with the experimental data. In order to determine the degree of compliance of the LiCl – LiNO₃ – H₂O system at 25 °C with Zdanovskii's rule, we

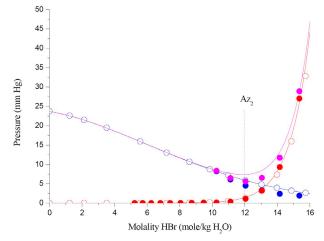


Fig. 2. The data calculated based on the Pitzer equations (lines and open circles) and the experimental data [17] (solid circles) regarding the concentration dependence of the pressures: H_2O (blue), HBr (red), and total (violet). Az_2 – azeotrope

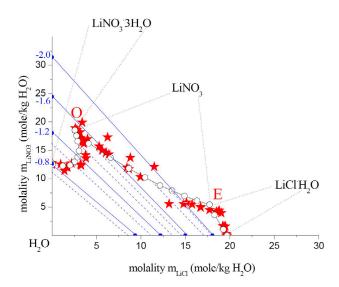


Fig. 3. Solubility diagram of the LiCl – LiNO $_3$ – $\rm H_2O$ system at 25 °C (open circles and lines – calculation results, red stars – experimental data [19], blue solid lines – calculated water isoactivates, $\rm lna_W$ values are shown in blue, blue dotted lines – experimental data obtained by the authors). According to the general classification of invariant points [20, 21], point O is a so–called through point, corresponding to the saturation with two different crystalline hydrates of one salt component, and point E is eutonic

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Table 3. Dependence of osmotic coefficients on the solution composition in the LiCl – LiNO $_3$ – $\rm H_2O$ system at 25 $^{\circ}C$

m_{LiNO_3} (mol/kg H_2O)	m_{LiCl} (mol/kg H ₂ O)	$\phi_{\rm H_2O}$	m_{LiNO_3} (mol/kg H ₂ O)	$m_{\rm LiCl}$ (mol/kg ${ m H_2O}$)	$\phi_{\rm H_2O}$
	$\ln a_W = -0.6999$		$\ln a_W = -1.0674$		
11.36	0.00	1.71	16.55	0.00	1.79
9.50	1.39	1.78	13.62	1.99	1.90
8.02	2.51	1.84	11.34	3.56	1.99
6.65	3.55	1.90	9.28	4.95	2.08
5.34	4.54	1.96	7.37	6.27	2.17
4.30	5.33	2.01	5.89	7.36	2.14
3.26	6.13	2.06	4.43	8.31	2.33
2.39	6.78	2.11	3.21	9.11	2.41
1.48	7.47	2.17	1.98	9.95	2.49
0.65	8.10	2.22	1.71	10.11	2.56
0.00	8.59	2.26	0.00	11.29	2.60
	$\ln a_W = -1.3490$			$\ln a_W = -1.5055$	
16.88	2.47	1.93	18.75	2.74	1.94
13.90	4.36	2.05	15.39	4.83	2.06
11.28	6.03	2.16	12.42	6.64	2.19
8.88	7.57	2.28	9.76	8.31	2.31
7.05	8.73	2.37	7.71	9.59	2.41
5.26	9.87	2.47	4.14	11.75	2.63
3.81	10.80	2.56	2.53	12.76	2.73
2.33	11.73	2.66	0.00	14.31	2.92
0.00	13.23	2.83		$\ln a_W = -1.8461$	
	$\ln a_W = -1.6741$		7.18	12.45	2.61
8.64	10.14	2.47	4.84	13.84	2.68
6.59	11.43	2.58	4.29	14.18	2.77
5.37	12.18	2.65	2.93	15.05	2.85
3.95	13.06	2.73	1.45	15.91	2.95
2.70	13.84	2.81	0.00	16.80	3.05
1.34	14.70	2.90			
0.00	15.54	2.99			
	$\ln a_w = -1.9976$				
4.63	15.31	2.78			
3.16	16.18	2.87			
1.56	17.13	2.97			
0.00	18.06	3.07			

presented the deviations of the water isoactivates from linearity $\Delta_{Zd} = 1 - m_{\text{LiNO}_3} / m_{\text{LiNO}_3}^0 - m_{\text{LiCl}} / m_{\text{LiCl}}^0$ as a function of the Janecke index of LiNO₃ – $Y_{\text{LiNO}_3} = m_{\text{LiNO}_3} / (m_{\text{LiNO}_3} + m_{\text{LiCl}})$ over the whole array of experimental data (Fig. 4). Fig. 4 shows that for all the 55 experimental points, except for one, deviations from Zdanovskii's rule are very insignificant (≤ 0.002), which demonstrates

the accuracy of the isopiestic experiment. The thermodynamic potentials of the solid phases required for the calculation of the solubility diagrams of the LiCl – LiNO $_3$ – $\rm H_2O$ system at 25 °C are presented in Table 4. The table also demonstrates the thermodynamic potentials – $\ln SP (SP - \text{solubility product of the solid phase})$ of all the other solid phases in the subsequent

Table 4. Thermodynamic potentials of solid phases (ln SP) at 25 °C

Solid phase	ln SP	Solid phase	ln SP	Solid phase	ln SP
NaCl	3.65	LiCl·H ₂ O	12.13	KCl	2.09
LiNO ₃	9.08	LiNO ₃ ·3H ₂ O	4.73		

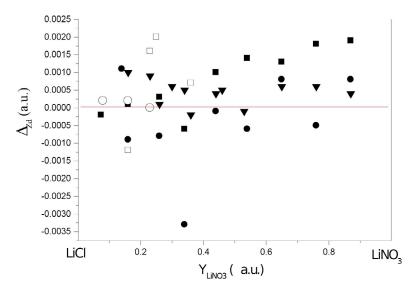


Fig. 4. Deviations from Zdanovskii's rule in the LiCl – LiNO $_3$ – H_2 O system at 25 °C: $Y_{\text{LiNO}_3} = m_{\text{LiNO}_3} / (m_{\text{LiNO}_3} + m_{\text{LiCl}})$; $\Delta_{Zd} = 1 - m_{\text{LiNO}_3} / m_{\text{LiNO}_3}^0 - m_{\text{LiCl}} / m_{\text{LiCl}}^0$. Designations: solid squares – $\ln a_W = -0.6999$; solid circles – $\ln a_W = -1.0674$; up triangles – $\ln a_W = -1.3490$; down triangles – $\ln a_W = -1.5055$; rhombi – $\ln a_W = -1.6741$; open squares – $\ln a_W = -1.8461$; open circles – $\ln a_W = -1.9976$

systems. The values of ln *SP* are calculated based on the solubility in the binary systems.

4. Ternary systems with two volatile components (HCl - KCl - H₂O, HCl - NaCl - H₂O) at 25 °C

The ternary systems with two volatile components considered in our study were HCl – KCl – H_2O and HCl – NaCl – H_2O at 25 °C. The existing literature presents a lot of experimental data regarding these systems [19–23]. The process of plotting of solubility diagrams (in our case, univariant crystallization curves of NaCl and KCl) is quite clear.

4.1. An algorithm for the calculation of solubility diagrams

A1) Define a_{w} .

A2) Find m_1^0 , m_2^0 in the binary subsystems corresponding to a_w . This can be done based on the experimental data or an accurate thermodynamic model, for instance the Pitzer model. Values m_1^0 , m_2^0 can correspond to both stable homogeneous solutions and supersaturated solutions. In the

latter case, it is necessary to extrapolate the model for supersaturated solutions.

A3) Determine the activity of salt components in binary solutions – a_1^0 , a_2^0 based on the experimental data or a thermodynamic model.

A4) Determine the thermodynamic potential of the solid phase $M_{v_{M1}}X_{v_x}$ based on the solubility of the salt – m_1^s :

$$\ln SP(M_{v_{M1}}X_{v_{X}}) = \ln a_{1}^{s} = v_{M1}\ln(v_{M1}m_{1}^{s}) + +v_{X}\ln(v_{X}m_{1}^{s}) + (v_{M1} + v_{X})\ln\gamma_{1}^{s},$$
(15.1)

where a_1^s and γ_1^s are the activity and activity coefficient of $M_{v_{M_1}}X_{v_{v}}$ in the saturated solution.

A5) Use equation (12.3), taking into account that $a_i^{(mult)} \equiv a_i^s$:

$$\ln Z_1 = \ln SP(M_{v_{M1}} X_{v_x}) - \ln a_1^0$$
 (15.2)

and find Z_1 and $Z_2 = 1 - Z_1$.

A6) Find a point on the crystallization curve $M_{v_{MI}}X_{v_x}$:

$$\ln m_i = \ln Z_i + \ln m_i^0 \quad (i = 1, 2).$$
 (15.3)

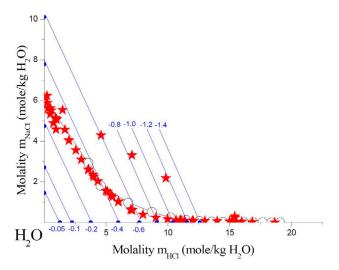


Fig. 5.1. Solubility diagram of the NaCl – HCl – $\rm H_2O$ system at 25 °C (open circles and lines – calculation results, red stars – experimental data [19, 20], blue lines – calculated water isoactivates, $\ln a_W$ values are shown in blue

The results of the calculation are presented in Fig. 5.1 and 5.2. The figures also demonstrate the isoactivates of water calculated based on Zdanovskii's rule and extrapolated to the region of metastable supersaturated solutions. Fig. 5 demonstrates that the experimental data is in good agreement with the results of the calculation based on the binary subsystems.

4.2. An algorithm for the calculation of solid-vapor-liquid equilibrium diagrams

Let's calculate the partial pressures of volatile components, i.e. the solvent (*W*) and HCl, above saturated solutions in both systems using the following algorithm.

- B1) Define a_w and calculate $\ln P_w = \ln a_w + \ln P_w^{(0)}$.
- B2) Calculate $\ln a_{\rm HCl}^0$, $Z_{\rm HCl}$ (see algorithm 4.1) and $P_{\rm HCl} = a_{\rm HCl}^0 Z_{\rm HCl} K_{\rm H,HCl}^m$.
- B3) Calculate the total pressure of volatile components in the system: $P_{sum} = P_W + P_{HCl}$.
- B4) The composition of the solution is still described by the equation (15.3):

$$\ln m_{\text{HCl}} = \ln Z_{\text{HCl}} + \ln m_{\text{HCl}}^{0};$$

$$\ln m_{\text{NaCl}} = \ln Z_{\text{NaCl}} + \ln m_{\text{NaCl}}^{0}.$$
(16)

The calculated pressures above saturated solutions in both systems are given in Fig. 6.1, 6.2 and Tables 5.1, 5.2. The figures demonstrate that in both systems a pseudo-azeotrope negative with regard to $P_{\rm T}$ is implemented with a minimum total

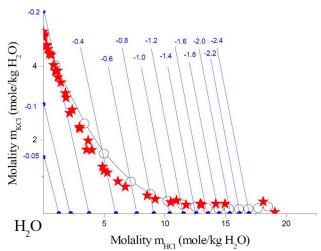


Fig. 5.2. Solubility diagram of the KCl – HCl – $\rm H_2O$ system at 25 °C (open circles and lines – calculation results, red stars – experimental data [19, 23, 24], blue lines – calculated water isoactivates, $\ln a_W$ values are shown in blue

pressure $-P_{Azi}$. The calculation based on the binary subsystems of the ternary $HCl - KCl - H_2O$ system at 25 °C are in good agreement with the only experimental data that we managed to find [24].

4.3. An algorithm for the calculation of vapor-liquid equilibrium diagrams in unsaturated solutions

C1) Obtain linear water isoactivates with $P_W = const$, using trivial methods (see Fig. 5.1, 5.2), including in the metastable region of supersaturated solutions.

C2) Curvilinear isoactivates or isobars of partial pressures $P_{\rm HCl}$ = const are obtained by solving equations of the following type: $a_{\rm HCl}^0 Z_{\rm HCl} K_{\rm H,HCl}^m = const$ with various $a_{\rm W}$, finding $Z_{\rm HCl}$ and the corresponding compositions of solutions ($m_{\rm HCl}$, $m_{\rm KCl}$), followed by combining the solutions to the equations in a continuous smooth curve.

C3) The curvilinear isobars of the total pressures of volatile components $P_{sum} = const$ are also obtained by solving equations of type $a_{\rm HCl}^0 Z_{\rm HCl} K_{\rm H,HCl}^m + P_W = const$ with various a_W and finding the corresponding $Z_{\rm HCl}$ and the corresponding compositions of solutions ($m_{\rm HCl}$), followed by combining the solutions to the equations in a continuous smooth curve. The latter curves, i.e. isobars of the total pressure in both ternary systems, are given in Fig 7.1 and 7.2. In both figures, some of the vapor-liquid

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-2.0

-2.2

3.21

2.63

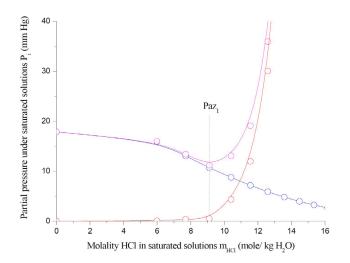
329.2

663.4

332.4

666.0

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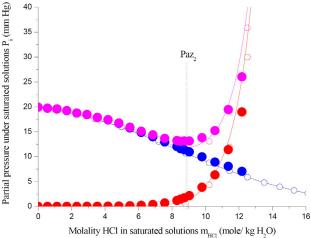


Fig. 6.1. Dependence of partial pressures $P_{\rm H_2O}$ (blue lines and circles), $P_{\rm HCI}$ (red lines and circles), and the total pressure P_{sum} (violet lines and circles) in the NaCl – HCl – H₂O system at 25 °C – calculation based on binary subsystems

Fig. 6.2. Dependence of partial pressures $P_{\rm H_2O}$ (blue lines and circles), $P_{\rm HCI}$ (red lines and circles), and the total pressure $P_{\rm sum}$ (violet lines and circles) in the KCl – HCl – H₂O system at 25 °C – calculation based on binary subsystems, solid circles – experimental data [24]

Table 5.1 Solubility, partial, and total pressures in saturated solutions in the HCl – NaCl – H_2O system at 25 °C

$\ln a_{_W}$	$m_{ m HCl}^0$	$m_{ m NaCl}^{ m 0}$	$\ln a_{\scriptscriptstyle ext{HCl}}^{\scriptscriptstyle 0}$	$\ln a_{ ext{NaCl}}^0$	$Z_{ m _{NaCl}}$	$Z_{\scriptscriptstyle ext{HCl}}$	$m_{_{ m NaCl}}$	$m_{_{ m HCl}}$
r.u.	mol/	kg H₂O	1	.u.	r.u.		mol/kg H ₂ O	
0.00	0.00	0.00	_	3.65	1.00	0.00	6.16	0.00
-0.05	1.28	1.45	0.19	-0.10	_	_	_	_
-0.1	2.26	2.71	1.78	1.27	_	_	_	_
-0.2	3.77	4.74	3.64	2.78	_	_	_	_
-0.4	5.99	7.78	5.93	4.56	0.4025	0.597	3.131	3.57
-0.6	7.70	10.11	7.55	5.80	0.1164	0.883	1.177	6.80
-0.8	9.11	12.05	7.87	6.81	0.0424	0.957	0.511	8.72
-1.0	10.41	13.74	10.00	7.67	0.0179	0.982	0.246	10.22
-1.2	11.56	15.24	11.01	8.43	0.0084	0.991	0.127	11.46
-1.4	12.60	16.50	11.93	9.13	0.0041	0.995	0.068	12.54
-1.6	13.58	17.85	12.78	9.78	0.0022	0.997	0.038	13.55
-1.8	14.49	19,01	13.57	10,38	0.0012	0.998	0.022	14.47
-2.0	15.35	20,09	14.32	10,95	6.75.10-4	0.999	0.013	15.33
-2.2	16.16	21,11	15.02	11,48	3.97.10-4	0.9996	0.0083	16.15
$\ln a_{_W}$	$P_{_W}$	$P_{_{ m HCl}}$	P_{sum}					
r.u.		mm Hg						
-0.2850	17.86	0.000	17.86					
-0.4	15.92	0.044	15.97					
6	13.03	0.333	13.37					
-0.8	10.67	0.498	11.17					
-1.0	8.74	4.30	13.04					
-1.2	7.15	11.92	19.08					
-1.4	5.85	30.04	35.90					
-1.6	4.79	70.44	75.23					
-1.8	3.92	155.3	159.2					

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Table 5.2. Solubility, partial, and total pressures in saturated solutions in the HCl – KCl – H₂O system at 25 °C

at 25 G								
$\ln a_w$	$m_{ m HCl}^0$	$m_{ m KCl}^0$	$\ln a_{\text{HCl}}^{0}$	$\ln a_{\text{KCl}}^{0}$	$Z_{ ext{ iny KCl}}$	$Z_{\scriptscriptstyle ext{HCl}}$	$m_{_{ m KCl}}$	$n_{_{ m HCl}}$
r.u.	mol/	kg H ₂ O		r.u.	r.u	r.u.		g H ₂ O
0.00	0.00	0.00	0.00	2.08	1.000	0.000	4.82	0.00
-0.05	1.28	1.53	0.19	-0.23	_	_	_	_
-0.1	2.26	2.97	1.78	1.04	_	_	_	_
-0.2	3.77	5.49	3.64	2.39	0.733	0.266	4,02	1.00
-0.3	4.97	7.72	4.91	3.23	0.316	0.683	2,44	3.39
-0.4	5.99	9.75	5.93	3.87	0.166	0.833	1.62	4.98
-0.6	7.70	13.49	7.55	4.83	0.063	0.936	0.86	7.20
-0.8	9.11	16.97	7.87	5.56	0.030	0.969	0.52	8.82
-1.0	10.41	20.38	10.00	6.16	0.016	0.983	0.34	10.23
-1.2	11.56	23.83	11.01	6.67	0.010	0.989	0.24	11.44
-1.4	12.60	27.48	11.93	7.10	0.0066	0.993	0.18	12.51
-1.6	13.58	31.57	12.78	7.47	0.0045	0.995	0.14	13.51
-1.8	14.49	36.85	13.57	7.80	0.0032	0.996	0.12	14.44
-2.0	15.35	41.83	14.31	7.96	0.0027	0.997	0.12	15.30
-2.2	16.16	47.43	15.02	8.09	0.0024	0.997	0.12	16.12
$\ln a_{\scriptscriptstyle W}$	$P_{_W}$	$P_{\scriptscriptstyle ext{HCl}}$	P_{sum}					
r.u.		mm Hg						
-0.1720	19.90	0.00000	19.90					
-0.2	19.45	0.00202	19.45					
-0.3	17.60	0.0184	17.62					
-0.4	15.92	0.0625	15.98					
-0.6	13.03	0.354	13.39					
-0.8	10.67	0.504	11.18					
-1.0	8.74	4.308	13.04					
-1.2	7.15	11.90	19.05	_				
-1.4	5.85	29.98	35.84					
		1	1	1				

diagrams correspond to stable solutions, and some (with high salt concentrations) correspond to metastable solutions supersaturated with respect to salts. Both figures demonstrate a univariant line of local extrema of function P_{sum} (in our case, local minimums), i.e. a negative fold on the surface of $P_{sum}(m_{HCl}, m_{salt})$ [7, 25, 26]. This fold lies between the binary azeotrope – Az₁ and the salt-water binary subsystem, where it degrades, because in binary systems dependence

70.30

155.0

325.3

661.6

75.09

158.9

328.5

664.3

4.79

3.92

3.21

2.63

-1.6

-1.8

-2.0

-2.2

$$P_{sum}(m_{salt})$$
 is strictly monotonous $\frac{dP_{sum}}{dm_{salt}} < 0$ and

cannot have extrema, according to the criteria of diffusion resistance: [7–9].

5. Quaternary systems with two volatile components (HCl - NaCl - KCl - H_2O) at 25 °C

The calculation of phase equilibria in quaternary systems with two volatile components was performed using the HCl(1) - NaCl(2) - $KCl(3) - H_2O$) system at 25 °C.

5.1. An algorithm for the calculation of solubility diagrams

D1) Define value a_W . D2) Find m_1^0 , m_2^0 , m_3^0 in the binary subsystems corresponding to a_w . This can be done based on the experimental data or an accurate thermodynamic

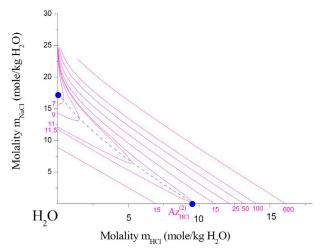


Fig. 7.1. Dependence of the total pressure of volatile components in the NaCl – HCl – $\rm H_2O$ system at 25 °C: solid violet lines – calculation based on binary subsystems; P_{sum} values are represented by digits of the same color. The dotted blue line is the line of local extrema of the P_{sum} function (in our case, local minima), i.e. a negative fold on the surface of $P_{sum}(m_{\rm HCl}, m_{\rm NaCl})$ [7, 8]. The blue dots of the binary azeotrope in the HCl – $\rm H_2O$ subsystem are below and the point of degeneration of the fold is on the left

model for instance the Pitzer model. Values m_1^0, m_2^0, m_3^0 can correspond to both stable homogeneous solutions and supersaturated solutions. In the latter case, it is necessary to extrapolate the model for supersaturated solutions.

D3) Determine the activity of salt components in binary solutions $-a_1^0, a_2^0, a_3^0$ based on the experimental data or a thermodynamic model.

D4) Determine the thermodynamic potentials of two solid phases i = 2, 3 (NaCl, KCl) crystallized on a single crystallization curve on the solubility diagram of the quaternary system based on the solubility of the salts $-m_i^s$: $\ln SP_i$.

D5) Use equation (12.3), taking into account that $a_i^{(mult)} \equiv a_i^s$:

$$ln Z_i = ln SP_i - ln a_i^0$$
(17.1)

and find Z_i (i = 2, 3) and $Z_1 = 1 - Z_2 - Z_3$.

D6) Find a point on the univariant crystallization curve of phases 2 and 3:

$$\ln m_i = \ln Z_i + \ln m_i^0 \ (i = 1, 2, 3).$$
 (17.2)

The results of the calculation of the solubility diagram of the quaternary system are given in

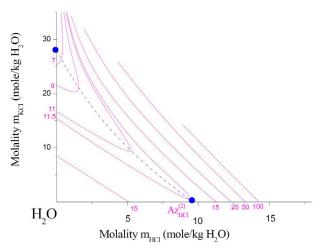


Fig. 7.2. Dependence of the total pressure of volatile components in the KCl – HCl – $\rm H_2O$ system at 25 °C: solid violet lines – calculation based on binary subsystems; P_{sum} values are represented by digits of the same color. The dotted blue line is the line of local extrema of the P_{sum} function (in our case, local minima), i.e. a negative fold on the surface of P_{sum} ($m_{\rm HCl}$, $m_{\rm KCl}$) [7, 8]. The blue dots of the binary azeotrope in the HCl – $\rm H_2O$ subsystem are below and the point of degeneration of the fold is on the left

Fig. 8.1 and Table 6. The results are presented in comparison with the experimental data provided in [24], which are in good agreement.

5.2. An algorithm for the calculation of solid₁-solid₂-liquid-vapor equilibrium diagrams

Let's calculate the partial pressures of volatile components, i.e. the solvent (W) and HCl, above saturated solutions in the HCl – NaCl – KCl – $\rm H_2O$ system at 25 °C. The following algorithm can be used.

E1) Define a_w and calculate $\ln P_w = \ln a_w + \ln P_w^{(0)}$.

E2) Calculate $\ln a_{\rm HCl}^0$, $Z_{\rm HCl}$ (see algorithm 5.1) and $P_{\rm HCl} = a_{\rm HCl}^0 Z_{\rm HCl} K_{\rm H,HCl}^m$.

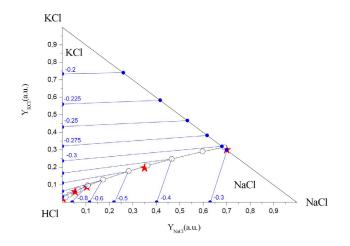
E3) Calculate the total pressure of volatile components in the system: $P_{yy} = P_{yy} + P_{yy}$.

components in the system: $P_{sum} = P_W + P_{HCI}$. E4) The composition of the solution is still described by the equation (17.2):

The calculated pressures above saturated solutions in both systems are given in Fig. 8.2 and Table 6. Fig. 8.2 demonstrates that a pseudo-azeotrope negative with respect to P_T is implemented with a minimum total pressure – P_{Az3} .

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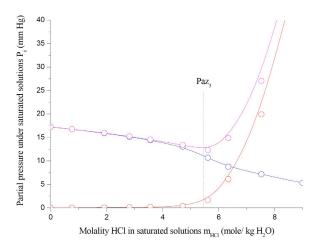


Fig. 8.1. Solubility diagram of the KCl – NaCl – HCl – H_2O system at 25 °C (open circles and lines – calculation results, red stars – experimental data [38], blue lines – calculated water isoactivates, $\ln A_W$ values are shown in blue **Fig. 8.2.** Dependence of partial pressures P_{H_2O} (blue lines and circles), P_{HCl} (red lines and circles), and the total pressure P_{Sum} (violet lines and circles) in the KCl – NaCl – HCl – H_2O system at 25 °C – cal-

culation based on binary subsystems

Table 6. Solubility, partial, and total pressures in saturated solutions in the NaCl-KCl-HCl- H₂O system at 25 °C

$ln a_w$	$m_{ m NaCl}^0$	$m_{ m KCl}^{ m 0}$	$m_{ m HCl}^0$	$\ln a_{ ext{NaCl}}^0$	$\ln a_{ ext{KCl}}^0$	$Z_{ ext{NaCl}}$	$Z_{\scriptscriptstyle m KCl}$	$Z_{\scriptscriptstyle ext{HCl}}$	$m_{_{ m NaCl}}$	$m_{_{ m KCl}}$	$m_{_{ m HCl}}$
r.u.	m	ol/kg H ₂	0		r.u.				m	O	
-0.32	6.69	8.14	5.18	3.94	3.37	0.728	0.272	0.000	5.00	2.28	0,00
-0.35	7.11	8.76	5.49	4.19	3.57	0.582	0.229	0.187	414	2.01	0.77
-0.40	7.78	9.75	5.99	4.56	3.87	0.402	0.170	0.427	3.13	1.66	1.92
-0.45	8.41	10.72	6.45	4.90	4.14	0.286	0.130	0.583	2.40	1.39	2.82
-0.50	9.01	11.66	6.89	5.22	4.39	0.208	0.101	0.690	1.87	1.18	3.57
-0.60	10.12	13.48	7.70	5.80	4.83	0.116	0.065	0.818	1.17	0.87	4.72
-0.80	11.12	15.24	9.14	6.32	5.22	0.069	0.044	0.886	0.77	0.67	5.62
-1.00	12.05	16.97	10.41	6.81	5.56	0.042	0.031	0.926	0.51	0.53	6.35
-1.20	13.74	20.37	11.55	7.67	6.16	0.017	0.017	0.964	0.24	0.35	7.53

$\ln a_{_W}$	$P_{\scriptscriptstyle W}$	$P_{\scriptscriptstyle ext{HCl}}$	P_{sum}
r.u.		mm Hg	
-0.32	17.25	0.000	17.25
-0.35	16.74	0.0090	16.75
-0.40	15.92	0.033	15.96
-0.45	15.15	0.073	15.22
-0.50	14.41	0.134	14.54
-0.60	13.03	0.359	13.39
-0.80	10.66	1.64	12.31
-1.00	8.74	6.11	14.85
-1.20	7.15	19.90	27.06

6. Topological isomorphism of *n*-phase equilibrium diagrams of *n*-component systems with two volatile components in decreased concentration spaces

This section focuses on the topological isomorphism of equilibrium diagrams of multicomponent multiphase systems containing two volatile components in univariant multicomponent multiphase systems. Earlier studies considered some particular systems of the same or similar types (for instance, [27]).

6.1. Topological isomorphism of three-phase equilibrium diagrams of ternary systems with two volatile components

Let's consider a three-phase equilibrium (solid phase – non-volatile component 3 (s) – vapor, consisting of two volatile components 1 – volatile acid 2 – solvent (v) – liquid solution, consisting of all the three components of the ternary system) at a constant temperature T = const. For instance, let's consider the following system: HCl(1) – H₂O(2) – NaCl(3). Let's introduce the partial Gibbs energy:

$$G^{[3]} = G - \mu_3 n_3 = \sum_{i=1}^{2} \mu_i n_i, \qquad (18)$$

where μ_i and n_i are the chemical potential and the number of moles of the *i*-th volatile component.

Let's consider a heterogeneous ternary system and introduce a scalar quantity $y_1^{(i)}$, where $y_1^{(i)}$ is the mole fraction of the 1st component (in the i-th phase) in the concentration space reduced with respect to the 3rd component. These are the so-called Janecke indices:

$$y_1 = n_1 / \sum_{i=1}^{2} n_j. {19}$$

The independent characteristic parameters of the partial molar Gibbs energy (18) are as follows: temperature -T, pressure -P, the Janecke index $-y_1$, and the chemical potential of the 3rd component in the solid phase $-\mu_3$. The equation of the two-phase $(\alpha-\beta)$ equilibrium shift can be described by the following system of generalized differential van der Waals equations:

$$(y_1^{(\alpha)} - y_1^{(\beta)})G_{11}^{[3](\alpha)}dy_1^{(\alpha)} = = S^{[3](\alpha \to \beta)}dT - V^{[3](\alpha \to \beta)}dP + n_{\bar{s}}^{[3](\alpha \to \beta)}d\mu_{\bar{s}},$$
 (20)

$$(y_1^{(\beta)} - y_1^{(\alpha)}) G_{11}^{[3](\beta)} dy_1^{(\beta)} = = S^{[3](\beta \to \alpha)} dT - V^{[3](\beta \to \alpha)} dP + n_5^{[3](\beta \to \alpha)} d\mu_3,$$
 (21)

where:

$$\begin{split} S^{\left[3\right](\alpha\to\beta)} &= \left[S^{\left[3\right](\beta)} - S^{\left[w\right](\alpha)} + (y_1^{(\beta)} - y_1^{(\alpha)})\nabla S^{\left[3\right](\alpha)}\right] \\ S^{\left[3\right](\beta\to\alpha)} &= \left[S^{\left[3\right](\alpha)} - S^{\left[3\right](\beta)} + (y_1^{(\alpha)} - y_1^{(\beta)})\nabla S^{\left[3\right](\beta)}\right] \\ V^{\left[3\right](\alpha\to\beta)} &= \left[V^{\left[3\right](\beta)} - V^{\left[3\right](\alpha)} + (y_1^{(\beta)} - y_1^{(\alpha)})\nabla V^{\left[3\right](\alpha)}\right] \\ V^{\left[3\right](\beta\to\alpha)} &= \left[V^{\left[3\right](\alpha)} - V^{\left[3\right](\beta)} + (y_1^{(\alpha)} - y_1^{(\beta)})\nabla V^{\left[3\right](\beta)}\right] \\ n_w^{\left[3\right](\alpha\to\beta)} &= \left[n_w^{\left[3\right](\beta)} - n_w^{\left[3\right](\alpha)} + (y_1^{(\beta)} - y_1^{(\alpha)})\nabla n_3^{\left(3\right)(\alpha)}\right] \\ n_w^{\left[3\right](\beta\to\alpha)} &= \left[n_w^{\left[3\right](\alpha)} - n_w^{\left[3\right](\beta)} + (y_1^{(\alpha)} - y_1^{(\beta)})\nabla n_w^{\left(3\right)(\beta)}\right], \end{split}$$

where $V^{[3](\tau)}$, $S^{[3](\tau)}$ and $n_w^{[3](\tau)}$ are the molar volume, entropies, and numbers of moles of the 3rd component of phase τ , calculated without taking into account the number of moles of the salt (per 1 mole of volatile components). The $\nabla V^{[3](\tau)}$, $\nabla S^{[3](\tau)}$, $\nabla n_w^{[3](\tau)}$ functions are gradients of concentration of the corresponding functions in the reduced concentration space, $\left(\partial V^{[3](\tau)} / \partial y_1^{(\tau)}\right)_{T,P,u_v^{(\tau)}}$, $\left(\partial S^{[3](\tau)} / \partial y_1^{(\tau)}\right)_{T,P,u_v^{(\tau)}}$, and

$$\left(n_w^{[\mathfrak{I}](\mathfrak{r})} / \partial y_1^{(\mathfrak{r})}\right)_{T,P,\mu_\mathfrak{I}^{(\mathfrak{r})}}$$
, respectively; $y_1^{(\mathfrak{r})}$ is the param-

eter, characterizing the state of the figurative point of phase τ in the reduced concentration space; $dy_1^{(\tau)}$ characterizes the displacement of $y_1^{(\tau)}$ according to the displacement of the two-phase equilibrium; $G_{11}^{[3](\tau)}$ is determined as:

$$G_{11}^{[3](\alpha)} = \left(\frac{\partial^2 G^{[1](\alpha)}}{\partial y_1^{(\alpha)2}}\right)_{T = p_{11}(\alpha)}$$
(23)

and

$$G_{11}^{[\bar{3}](\beta)} = \left(\frac{\partial^2 G^{[1](\beta)}}{\partial y_1^{(\beta)2}}\right)_{T,P,\mu_3^{(\beta)}}.$$
 (24)

According to the diffusion resistance criteria and Sylvester's criterion, for the positive definite quadric forms, the following is true:

$$G_{11}^{[3](\tau)} > 0$$
 . (25)

The $S^{[3](\alpha \to \beta)}$, $S^{[3](\beta \to \alpha)}$, $V^{[3](\alpha \to \beta)}$, $V^{[3](\beta \to \alpha)}$, $n_w^{[3](\beta \to \alpha)}$ parameters have the physical senses of entropy, the volume and the number of moles of the 3rd

component in the isotherm–isobar–isopotential (with respect to the 3rd component) process of formation of one mole of phase β from an infinitely large.

Let's assume that the α phase is a liquid solution (l), the β phase is vapor (v), and the $\alpha \to \beta$ equilibrium is (l)–(v), dT = 0, $d\mu_3$ = 0. The latter is true because the standard chemical potential of the solid phase (s), μ_3 , depends on the temperature only. Therefore, we are dealing with a three-phase equilibrium (l)–(v)–(s). Let's rewrite the system of differential equations (20, 21):

$$(y_1^{(l)} - y_1^{(v)})G_{11}^{[3](l)}dy_1^{(l)} = -V^{[3](l \to v)}dP$$
(26)

$$(y_1^{(v)} - y_1^{(l)})G_{11}^{[3](v)}dy_1^{(v)} = -V^{[3](v \to l)}dP . (27)$$

Naturally, the equation cannot be used with dP = 0, $dT \neq 0$, because then $d\mu_3 \neq 0$, and the corresponding term should be added to equations (26, 27).

Based on the physical sense, $V^{[3](l\to v)}>0$, $V^{[3](v\to l)}<0$, and therefore we can formulate three analogues of the three Gibbs–Konovalov and Gibbs–Roozeboom rules.

First rule.

When moving along the crystallization curve of the non-volatile component on the isothermal solubility diagram of a ternary system, the total pressure of vapors above saturated solutions increases (decreases) following an increase in the Janecke index of the volatile component (excluding the non-volatile component), whose concentration in the ternary system of vapor is larger (smaller) than in the solution:

$$\left[dP/dy_1^{(l)}\right]_{T_{11}(x)} > (<)0$$
, if $y_1^{(v)} > (<)y_1^{(l)}$. (28.1)

Second rule.

When moving along the crystallization curve of the non-volatile component on the isothermal solubility diagram of a ternary system in a reduced (excluding the non-volatile component) concentration space, the total pressure of the volatile components passes through the extremum, if the compositions of the existing phases in the reduced concentration space, i.e. the Janecke indices, coincide. These points will be called pseudo-azeotropes - $P_{\rm Az}$.

$$\left[dP/dy_1^{(l)}\right]_{T_{11}(x)} = 0$$
, if $y_1^{(v)} = y_1^{(l)}$. (28.2)

Third rule.

Dividing equation (26) by equation (27) we obtain:

$$\begin{bmatrix} dy_1^{(l)} / dy_1^{(v)} \end{bmatrix}_{T,\mu_3^{(\tau)}} =
= - \left[G_{11}^{[3](v)} / G_{11}^{[3](v)} \right] \left[V^{[3](l \to v)} / V^{[3](v \to l)} \right] > 0.$$
(28.3)

When moving along the crystallization curve of the non-volatile component on the isothermal solubility diagram of a ternary system in a reduced (excluding the non-volatile component) concentration space, the Janecke indices of the equilibrium liquid and vapor phases change simbatically.

6.2. Topological isomorphism of four-phase equilibrium diagrams of four-component systems with two volatile components

Let's consider a four-phase equilibrium (solid non-volatile component 3 (s_1) – solid non-volatile component 4 (s_2) – vapor, consisting of two volatile components 1 – volatile acid 2 and solvent (v) – liquid solution, consisting of all the four components of the four-component system) at a constant temperature T = const. For instance, let's consider the following system: HCl(1) – $H_2O(2)$ – NaCl(3) – KCl(4). Let's introduce the partial Gibbs energy:

$$G^{[3,4]} = G - \mu_3 n_3 - \mu_4 n_4 = \sum_{i=1}^{2} \mu_i n_i.$$
 (29)

Let's consider a heterogeneous four-component system and introduce a scalar quantity in the twice reduced concentration space $y_1^{(i)}$, where $y_1^{(i)}$ is the molar fraction of the 1-st component (in the i-th phase) in the twice reduced concentration space (excluding components 3 and 4). In this case, the Janecke indices are set by the following relation:

$$y_1 = n_1 / \sum_{j=1}^2 n_j \ . \tag{30}$$

The characteristic parameters of the partial molar Gibbs energy (29) are the following: temperature – T, pressure – P, the Janecke index – y_1 , and the chemical potentials of salts – μ_3 , μ_4 . The equation of the two-phase (α – β) equilibrium shift can be described by the following system of differential van der Waals equations:

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$$(y_1^{(\alpha)} - y_1^{(\beta)}) G_{11}^{[3,4](\alpha)} dy_1^{(\alpha)} = S^{[3,4](\alpha \to \beta)} dT - V^{[3,4](\alpha \to \beta)} dP + n_3^{[3,4](\alpha \to \beta)} d\mu_3 + n_4^{[3,4](\alpha \to \beta)} d\mu_4,$$
 (31)

$$(y_1^{(\beta)} - y_1^{(\alpha)}) G_{11}^{[3,4](\beta)} dy_1^{(\beta)} = S^{[3,4](\beta \to \alpha)} dT - V^{[3,4](\beta \to \alpha)} dP + n_3^{[3,4](\beta \to \alpha)} d\mu_3 + n_4^{[3,4](\beta \to \alpha)} d\mu_4,$$
 (32)

where parameters $G_{11}^{[3,4](\tau)}$, $S^{[3,4](\alpha \to \beta)}$, $S^{[3,4](\beta \to \alpha)}$, $V^{[3,4](\beta \to \alpha)}$, $N_3^{[3,4](\beta \to \alpha)}$, $N_4^{[3,4](\beta \to \alpha)}$, $N_5^{[3,4](\beta \to \alpha)}$, and $N_4^{[3,4](\beta \to \alpha)}$ have the same sense as the parameters in equations (22) with the only addition – $\mu_4^{(\tau)}$ should be constant for the corresponding derivatives:

$$\begin{split} &\left(\partial V^{[\mathfrak{I}](\tau)} \, / \, \partial y_{1}^{(\tau)}\right)_{T,P,\mu_{3}^{(\tau)},\mu_{4}^{(\tau)}}, \, \left(\partial S^{[\mathfrak{I}](\tau)} \, / \, \partial y_{1}^{(\tau)}\right)_{T,P,\mu_{3}^{(\tau)},\mu_{4}^{(\tau)}}, \\ &\left(n_{w}^{[\mathfrak{I}](\tau)} \, / \, \partial y_{1}^{(\tau)}\right)_{T,P,\mu_{3}^{(\tau)},\mu_{4}^{(\tau)}}, \, G_{11}^{[\mathfrak{I}](\tau)} = &\left(\frac{\partial^{2} G^{[\mathfrak{I}](\tau)}}{\partial y_{1}^{(\tau)2}}\right)_{T,P,\mu_{3}^{(\tau)},\mu_{4}^{(\tau)}}. \end{split}$$

Parameters $S^{[3](\alpha \to \beta)}$, $S^{[3](\beta \to \alpha)}$, $V^{[3](\alpha \to \beta)}$, $n_w^{[3](\alpha \to \beta)}$, and $n_w^{[3](\alpha \to \beta)}$ reflect the changes in the entropy, volume, and the number of moles of the components (3, 4) during the isotherm—isobarisopotential (with regard to the 3rd and 4th components) process of formation of one mole of phase β from an infinitely large mass of phase α , and vice versa in the concentration space, reduced with respect to the 3rd and 4th components.

Let's assume that the α phase is a liquid solution (l), the β phase is vapor (v), and the $\alpha \to \beta$ equilibrium is (l)–(v), dT = 0, d μ_3 = d μ_4 = 0. The latter is true because the standard chemical potentials of the components in the solid phase (s₁, s₂) μ_3 and μ_4 depend on the temperature only. Therefore, we are dealing with a fourphase equilibrium (l)–(v)–(s₁)–(s₂). Let's rewrite equations (31, 32) for the set conditions. Then we can easily derive equations (26, 27).

Therefore, we can formulate three analogues of the Gibbs–Konovalov and Gibbs–Roozeboom rules, which will be true for the twice reduced concentration space:

$$\left[dP/dy_1^{(l)}\right]_{T_{11}(\tau)_{11}(\tau)} > (<)0$$
, if $y_1^{(v)} > (<)y_1^{(l)}$; (33.1)

$$\left[dP / dy_1^{(l)} \right]_{T, \mathbf{u}_1^{(\tau)}, \mathbf{u}_1^{(\tau)}} = 0 , \text{ if } y_1^{(v)} = y_1^{(l)}; \tag{33.2}$$

$$\begin{split} & \left[dP / dy_{1}^{(l)} \right]_{T,\mu_{3}^{(\tau)},\mu_{4}^{(\tau)}} = \\ & = - \left[G_{11}^{\left[\overline{3} \right](v)} / G_{11}^{\left[\overline{3} \right](v)} \right] \left[V^{\left[\overline{3} \right](l \to v)} / V^{\left[\overline{3} \right](v \to l)} \right] > 0. \end{split} \tag{33.3}$$

The wording of the analogues is the same as that of the ones presented in section 6.1, the only difference being the following: "When moving along the **co-crystallization** curve of the **two** non-volatile components on the isothermal solubility diagram of a **four-component** system in a **twice** reduced (excluding both non-volatile components) concentration space...".

6.3. Topological isomorphism of n-phase equilibrium diagrams of n-component systems with two volatile components

Naturally, the suggested analogues of the three Gibbs-Konovalov and Gibbs-Roozeboom rules can be applied to n-phase heterogeneous systems within *n*-component systems with two volatile components. Let's consider a phase equilibrium (solid pure non-volatile component $3(s_1)$ – solid pure non-volatile component 4 (s_n) – solid pure non-volatile component $n(s_{n-1})$ – vapor, consisting of two volatile components 1 – volatile acid and 2 – solvent (v) – liquid solution, consisting of 1, 2, 3...n components) at a constant temperature T = const. In this case, we are dealing with an (n-2)-time reduced concentration space and can formulate another three analogues of the Gibbs-Konovalov and Gibbs-Roozeboom rules, which will be true for an (n-2)-time reduced concentration space:

$$\left[dP/dy_1^{(l)}\right]_{T,u_1^{(r)},u_1^{(r)},\dots,u_n^{(r)}} > (<)0$$
, if $y_1^{(v)} > (<)y_1^{(l)}$; (34.1)

$$\left[dP / dy_1^{(l)} \right]_{T,\mu_2^{(\tau)},\mu_4^{(\tau)},\dots,\mu_n^{(\tau)}} = 0 , \text{ if } y_1^{(\nu)} = y_1^{(l)}; \quad (34.2)$$

$$\begin{split} & \left[dP / dy_{1}^{(l)} \right]_{T,\mu_{3}^{(\tau)},\mu_{4}^{(\tau)},\dots,\mu_{n}^{(\tau)}} = \\ & = - \left[G_{11}^{\left[\bar{3}\right](v)} / G_{11}^{\left[\bar{3}\right](v)} \right] \left[V^{\left[\bar{3}\right](l \to v)} / V^{\left[\bar{3}\right](v \to l)} \right] > 0. \end{split}$$
 (34.3)

The wording of the analogues is the same as that of the ones presented in section 6.1, the only difference being the following: "When moving along the **co-crystallization** curve of (n-2) non-volatile components on the isothermal solubility diagram of an **n-component** system in an (n-2)-time reduced (excluding (n-2) non-volatile components) concentration space...".

6.4. Topological isomorphism of n-phase equilibrium diagrams of n-component systems with two volatile components with crystal solvates (crystalline hydrates) of solid components of constant composition

Until now, we assumed that all solid phases of constant composition that are in equilibrium with vapor and liquid solution are pure salt-type components. Let's assume that crystal solvates (crystalline hydrates) of solid solutions of constant compositions are crystallized in the system. Are they then the true analogues of the three Gibbs-Konovalov and Gibbs-Roozeboom rules presented in (34.1-34.3)? The answer to this question is positive provided that we consider crystalline hydrates or compounds of constant compositions rather than particular salts as independent components of the system. This can be done, because the independent components, or variables of the composition of a multicomponent system, are random, with only the number of independent components being constant. Therefore, the laws will be true in the concentration space reduced with respect to the new components. For instance, on the crystallization curve of crystalline hydrate $LiCl \cdot H_2O$ in the $HCl(1) - H_2O(2) - LiCl(3)$ system, all the analogues of the Gibbs-Konovalov and Gibbs-Roozeboom rules are true (28.1–28.3), if we recalculate the Janecke indices for the new components of the system: $HCl(1) - H_2O(2) -$

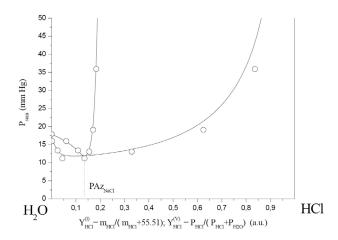


Fig. 9.1. Liquid-vapor phase diagram in the NaCl – $HCl - H_2O$ system at 25 °C in variables: the total vapor pressure of volatile components is the Yanecke HCl index; P_{AzNaCl} is a pseudo-azeotrope when moving along the NaCl crystallization curve

LiCl· $H_2O(3)$. In this case, the molar fractions in the new set of variables $n_i^{(new)}$ are easily calculated based on the molar fractions in the previous set of variables $n_i^{(old)}$: $n_{\text{LiCl-}H_2O}^{(new)} = n_{\text{LiCl}}^{(old)}$, $n_{\text{HCl}}^{(new)} = n_{\text{HCl}}^{(old)}$, $n_{\text{HCl}}^{(new)} = n_{\text{HCl}}^{(old)} - n_{\text{LiCl}}^{(old)}$. Let's consider another example: for the co-crystallization curve of $\text{MgCl}_2 \cdot \text{CaCl}_2 \cdot 12H_2O$ and crystalline hydrate $\text{MgCl}_2 \cdot 6H_2O$ in the $\text{HCl}(1) - H_2O(2) - \text{MgCl}_2(3) - \text{CaCl}_2(4)$ system the analogues of the said laws (28.1-28.3) are true, if we recalculate the molar fractions of the components with respect to $\text{HCl}(1) - H_2O(2) - \text{MgCl}_2 \cdot 6H_2O(3) - \text{MgCl}_2 \cdot \text{CaCl}_2 \cdot 12H_2O(4)$. The recalculation is performed based on the relations:

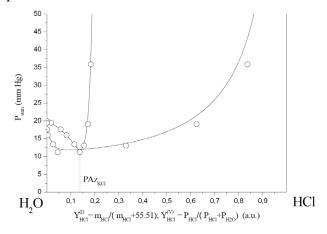


Fig. 9.2. Liquid-vapor phase diagram in the KCl – HCl – $\rm H_2O$ system at 25 °C in variables: the total vapor pressure of volatile components is the Yanecke HCl index; $P_{\rm AzKCl}$ is a pseudo-azeotrope when moving along the NaCl crystallization curve

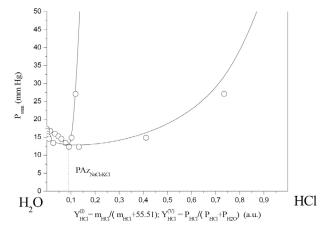


Fig. 9.3. Liquid-vapor phase diagram in the KCl – NaCl – HCl – $\rm H_2O$ system at 25 °C in variables: the total vapor pressure of volatile components is the Yanecke HCl index; $P_{\rm AzNaCl-KCl}$ is a pseudo-azeotrope when moving along the KCl and NaCl co-crystallization branch

$$\begin{split} n_{\rm MgCl_2 \cdot CaCl_2 \cdot 12H_2O}^{(new)} &= n_{\rm CaCl_2}^{(old)} \ , \quad n_{\rm MgCl_2 \cdot 6H_2O}^{(new)} = n_{\rm MgCl_2}^{(old)} - n_{\rm CaCl_2}^{(old)} \ , \\ n_{\rm HCl}^{(new)} &= n_{\rm HCl}^{(old)} \ , \qquad n_{\rm H_2O}^{(new)} = n_{\rm H_2O}^{(old)} - 6 \Big[n_{\rm MgCl_2}^{(old)} + n_{\rm CaCl_2}^{(old)} \Big] - \\ - 12 n_{\rm CaCl_2}^{(old)} \ . \end{split}$$

6.5. Application of the suggested analogues of the Gibbs-Konovalov and Gibbs-Roozeboom rules to ternary and quaternary systems

Fig. 9.1-9.3 demonstrate some examples of the suggested analogues of the Gibbs-Konovalov and Gibbs-Roozeboom rules implemented for univariant curves of three-phase equilibrium in ternary systems (HCl - H2O - NaCl and HCl -H₂O – KCl at 25 °C) and a quaternary system $(HCl - H_2O - NaCl - KCl at 25 °C)$. In the first case, three-phase equilibria (s-l-v) were considered, and in the second case - four-phase equilibria (s_1-s_2-l-v) . In the first case, the concentration space was reduced with respect to NaCl, and in the second case it was reduced with respect to NaCl and KCl. In both cases, the variables of the composition were the Janecke indices of the volatile components of the system – Y_{HCI} , Y_w . Fig. 9.1–9.3 show that the three systems are in complete agreement with the suggested analogues of the three rules. Each system has one pseudo-azeotrope – P_{Azi} (index j corresponds to the solid phase(s) with respect to which the considered solution is saturated). P_{Azj} complies with the minimum $P_{\text{sum}} = P_{W} + P_{\text{HCl}}$. All the diagrams are also limited from above by the boiling point of the solution at $P_{\text{sum}} = 760 \text{ mm Hg.}$

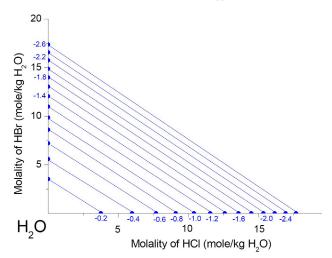


Fig. 10.1. Water isoacivates in the ternary HCl – HBr – H_2 O system at 25 °C, $\ln a_w$ values are shown in blue

7. Ternary systems with three volatile components (HCl - HBr - H $_2$ O) at 25 °C

To demonstrate the possibility of the calculation of phase diagrams of ternary systems with three volatile components, we considered the $HCl-HBr-H_2O$ system at 25 °C. Solid phases cannot be formed in such systems under the defined conditions. So, naturally, there is the only type of equilibrium diagram, which we consider below.

7.1. An algorithm for the calculation of vapor-liquid equilibrium diagrams in unsaturated solutions

F1) Define a_w .

F2) Obtain linear water isoactivates with $P_W = P_W^{(0)}$ ($a_W = const$) based on the data regarding the binary subsystems (see Fig. 10.1).

F3) Curvilinear isoactivates or isobars of partial pressures $P_{\rm HCl} = const$ are obtained by solving equations of the following type: $a_{\rm HCl}^0 Z_{\rm HCl} \, K_{\rm H,HCl}^m = const$ with various $a_{\rm W}$, finding $Z_{\rm HCl}$ and the corresponding compositions of solutions $(m_{\rm HCl}, m_{\rm KCl})$ (see (13)), followed by combining the solutions to the equations in a continuous smooth curve.

F4) Curvilinear isoactivates or isobars of partial pressures $P_{\rm HBr}$ = const are obtained by solving equations of the following type:

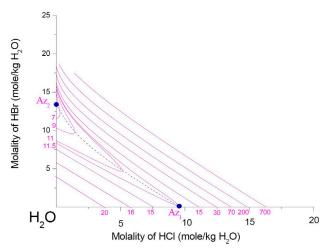


Fig. 10.2. Lines of equal total pressures of volatile components in the ternary HCl – HBr – H_2O system at 25 °C, P_{sum} values in mm Hg are shown in violet. The dotted blue line is the line of local extrema of the Psum function, i.e. a negative fold on the $P_{sum}(m_{HCl}, m_{HBr})$ surface [7, 8]; the blue dots are binary azeotropes in the subsystems HCl – H_2O and HBr – H_2O

 $a_{\rm HBr}^0 Z_{\rm HBr} K_{\rm H,HBr}^m = const$ with various $a_{\scriptscriptstyle W}$, finding $Z_{\rm HBr}$ and the corresponding compositions of solutions $(m_{\rm HCl}, m_{\rm KCl})$ (see (13)).

F5) Curvilinear isobars of the total pressures of volatile components $P_{\text{sum}} = const$ are obtained by solving equations of the following type: $a_{\text{HCl}}^0 Z_{\text{HCl}} K_{\text{H,HCl}}^m + a_{\text{HBr}}^0 \left(1 - Z_{\text{HCl}}\right) K_{\text{H,HBr}}^m + P_W = const$ with various a_W , finding the corresponding Z_{HCl} and the compositions of solutions $(m_{\text{HCl}}, m_{\text{KCl}})$, followed by combining the solutions to the equations in a continuous smooth curve. The latter curves, i.e. isobars of the total pressure in the considered ternary system, are given in Fig 10.2. Both figures demonstrate a univariant line of local extrema of function P_{sum} (local minimums), i.e. a negative fold on the surface of $P_{\text{sum}}(m_{\text{HCl}}, m_{\text{salt}})$ [7, 26, 27]. This fold lies between the binary azeotrope Az_1 and another binary azeotrope Az_2 , (see Fig. 10.2).

8. Conclusions

The article presented an algorithm for the calculation of vapor-liquid phase equilibria in multicomponent quasi-simple systems complying with Zdanovskii's rule. The algorithm is based entirely on the data regarding binary subsystems. The algorithm was tested using three- and four-component water-electrolyte systems with one, two, or three volatile components. The results of the calculations are in good agreement with the experimental data. Analogues of the Gibbs–Konovalov and Gibbs–Roozeboom rules were formulated and proved for the compositions of the solutions moving along univariant curves of n-phase equilibria in n-component systems at a constant temperature.

Author contributions

All authors made an equivalent contribution to the preparation of the publication.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Research article

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Analysis of the crystalline quality of bulk $In_{0.83}Ga(Al)_{0.17}$ layers formed on metamorphic InAlAs/InP buffer layers with linear and nonlinearcomposition gradients

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Abstract

This paper investigates the effectiveness of metamorphic InAlAs buffer layers with linear and root-like dependence of the In mole fraction in the composition for the growth of bulk $In_{0.83}Ga(Al)_{0.17}As$ layers on InP substrates. The analysis of the X-ray diffraction reciprocal space maps showed that in both cases $In_{0.83}Ga(Al)_{0.17}As$ layers were partially strain-free. One of the mechanisms of strain relaxation during the growth of the linearly graded buffer layer is the rotation of the crystal lattice, while the mechanism of strain relaxation during the growth of the convex-graded buffer layer is a 0.82° tilt of the crystal lattice without any rotation. According to the images obtained by transmission electron microscopy, the density of threading dislocations in the upper InGaAs layers grown on the buffer layer with a linear composition gradient is $\sim 5 \cdot 10^8$ cm⁻².

Keywords: Metamorphic buffer layers, Reciprocal space mapping of X-ray diffraction intensity, Transmission electron microscopy, Molecular beam epitaxy

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1. Introduction

The possibility of growing semiconductor layers on lattice-mismatched substrates has become an important factor contributing to the rapid development of epitaxial technologies. This achievement has allowed observing new electrophysical and optical effects in heterostructures and has considerably expanded the range of components for heterostructural nano- and optoelectronics. One of the traditional epitaxy-based approaches to obtaining semiconductor layers on substrates with a critically large mismatch in the crystal lattice parameters is the metamorphic growth. Metamorphic buffer layers (MBL) create a transition from the substrate lattice constant to the lattice constant required for the growth of the functional layers of the heterostructure. One way to enable such a transition is to gradually change the composition of the buffer layer, which means changing material fluxes during the MBL epitaxial growth. The change of the composition can be either smooth, for example, when a linear dependence is involved, or steplike. Typically, the thickness of the MBLs is a few micrometers. The MBL formation must result in a planar epitaxial surface for the subsequent strain-free growth of the active layers. The MBL technology is currently widely used for the manufacture of transistors with high mobility of electrons, lasers, solar cells, photodiodes, etc. [1– 6]. Various designs, thicknesses, and materials of buffer layers are studied and applied depending on the active layers of the heterostructure and its application in devices.

This paper is devoted to the study of possible designs of a metamorphic buffer layer for the growth of ${\rm In_{0.83}Ga_{0.17}As}$ triple solid solutions on InP substrates. These two materials are used for the manufacture of near-infrared spectral photodetectors, in which the peak wavelength of the photoresponse depends on the indium content in the photo-absorbing layer of InGaAs. The most common option is an absorbing layer of a ${\rm In_{0.53}Ga_{0.47}As}$ solid solution, which is lattice-matched with InP. The high quality of InP substrates, as well as the excellent characteristics of photodiodes and matrix photodetectors based on ${\rm In_{0.53}Ga_{0.47}As}/{\rm InP}$, allowed InGaAs to become the leading material for the manufacture of

photodetectors operating in the wavelength of 1–1.7 µm. With an increase in the indium content to the mole fraction of x = 0.83, the detection range shifts to a longer wavelength region of 2.2–2.6 µm. Thus, by varying the composition of the absorbing layer, a wider near-infrared region of the spectrum can be covered within the same system of materials. However, due to a large relative mismatch between the lattice parameters of $In_{0.83}Ga_{0.17}As$ and the InP substrate of ~ 2%, the critical thickness for the pseudomorphic growth of the active layer is 1-2 orders of magnitude less than the thickness required for the effective absorption of the emission [7]. Therefore, in order to avoid the appearance of a large number of misfit dislocations in the active region during the growth of the In_{0.83}Ga_{0.17}As layer on the surface of InP, it is necessary to grow transitional metamorphic buffer layers. In this case, MBLs contribute to the pseudomorphic growth of the InGaAs absorbing layers and, consequently, reduce the density of the resulting misfit dislocations, and also stop the penetration of threading dislocations into the active layers of the photodetectors. Dislocations have adverse effects on photodetectors since they contribute to some mechanisms of dark current formation, which leads to a reduced detectivity of devices [8].

2. Experimental

The research involved growing experimental samples containing MBLs of various designs by means of molecular-beam epitaxy using a Riber MBE49 unit (Fig. 1). The samples were InAlAs/ InGaAs heterostructures on n⁺-InP (100) "epiready" doped substrates. The heterostructures included a $\sim 0.1 \ \mu m$ thick layer of $In_{0.52}Al_{0.48}As$ which was lattice-matched with the substrate, InAlAs MBLs of ~ 2 µm of various configurations, an active region of In_{0.83}Ga_{0.17}As with a thickness of ~ 1.5 µm, and upper contact layers of $In_{0.83}Al(Ga)_{0.17}As$ with a thickness of ~ 650 nm (Fig. 1). The MBLs were grown at a constant temperature of the substrate, which was 70 °C lower than the growth temperature of the active region (490 °C). The process was concluded with a thermal cycling: the peak temperature rose to 530 °C followed by cooling to 100 °C. The buffer layers were doped with silicon to a level of 1·10¹⁸ cm⁻³ (n^+) , while the contact layers were doped with

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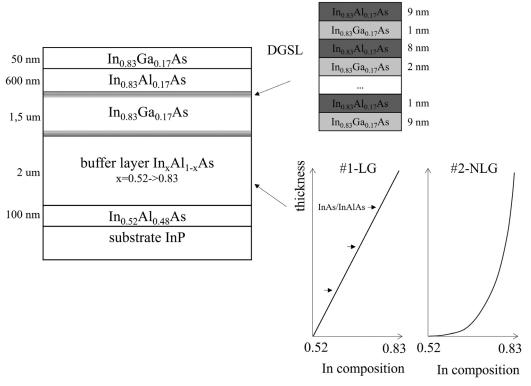


Fig. 1. Schematic image representing the layer composition of metamorphic heterostructures #1-LG μ #2-NLG

beryllium, with an impurity concentration of $2 \cdot 10^{18}$ cm⁻³ (p^+). The active region of InGaAs was also doped with silicon at a low concentration of $2 \cdot 10^{16}$ cm⁻³. The type and concentration of the dopant in each layer were selected in order to obtain a diode heterostructure for the subsequent manufacture of PIN photodiode crystals with an absorbing layer of In_{0.83}Ga_{0.17}As.

absorbing layer of $In_{0.83}Ga_{0.17}As$. Sample 1-LG contained $In_xAl_{1-x}As$ MBLs with a linear gradient of indium composition from x=0.52 to x=0.83 and with intermediate inserts of three InAs(4 Å)/InAlAs(10 Å) periods every 500 nm. Buffer layers of a similar design showed an advantage over the stepwise changing of the composition and the linear changing of the composition without inserts [9]. The demonstrated benefit of using this type of MBLs was in the lower density of threading dislocations in the functional layers and the existence of a dislocation-free region $d_{\rm free}$ in the buffer layer. The buffer layer of the 2-NLG sample was

The buffer layer of the 2-NLG sample was the In_xAl_{1-x}As layer whose composition changed following the nonlinear convex gradient which approximated the square root-like dependence. The fraction of InAs in the composition also varied from 0.52 to 0.83. The main difference between

the 1-LG and 2-NLG samples was that when the convex gradient of the composition was applied, the MBL crystal lattice parameter grew faster in its lower region bordering the substrate, and, conversely, more smoothly when approaching the active layers. Therefore, it can be assumed that dislocations formed more actively in the lower part of the MBLs, which was most distant from the functional layers of the heterostructure. Previous studies used metamorphic InAlAs layers on GaAs substrates to demonstrate the advantages of such a design as compared to a linear profile in terms of more effective elastic strain relaxation [10]. Also, using the example of the InAlAs/ GaAs system of materials, it was experimentally confirmed that the root-like dependence of the indium composition in MBLs resulted in a lower density of dislocations and the formation of a thicker d_{free} region [11].

To reduce the conduction band discontinuity between InGaAs and InAlAs and the dislocation density, the so-called digital-graded superlattices (DGSL) were introduced into the structure of the heterostructures from above and below the absorbing layer, which consisted of 9-period $In_{0.83}Ga_{0.17}As/In_{0.83}Al_{0.17}As$ superlattices with a

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total thickness of about 90 nm [12]. The wellto-barrier ratio in the DGSL period ranged from 1:9 to 9:1 in the case of the lower superlattice (MBL-active region heterointerface) and was mirrored in the case of the upper superlattice (active region-contact layer heterointerface). These superlattices created a stepped barrier for charge carriers at the point of conduction band discontinuity in the heterostructure. What is more, the additional heterointerfaces that DGSLs created were expected to facilitate a smoother transition from the lattice constant of InGaAs to InAlAs if there was any residual strain in InGaAs. Therefore, DGSL regions were inserted in both samples before the growth of the $In_{0.83}Ga_{0.17}As$ absorbing layer and after its complete formation.

The main experiments aimed at studying the crystalline properties of the samples of metamorphic heterostructures were preceded by their electrochemical capacitance–voltage (ECV) measurements. A comparison of the profiles of the distribution of the charge carriers concentration in the 1-LG and 2-NLG samples along the depth of the structure is shown in Fig. 2. The profile of the 1-LG sample was smoother and more comparable to the structure of the heterostructure in terms of layer thickness. It also did not have spikes and abrupt changes in concentration, unlike the 2-NLG sample. This may be evidence of a higher defectiveness of the 2-NLG sample with the root profile of the buffer layer composition. Based on the obtained results of the ECV measurements,

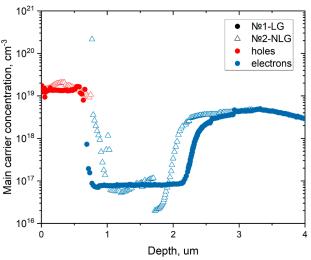


Fig. 2. Main carrier concentration profiles, obtained by electrochemical capacitance-voltage profiling

the 1-LG sample with a linear MBL was selected as a reference for the study by the destructive transmission electron microscopy (TEM). In addition, X-ray reflection spectra were measured for both samples, and reciprocal space maps of X-ray diffraction intensity were constructed and analyzed.

3. Results and discussion

3.1. Transmission electron microscopy

The 1-LG sample was examined by SEM in the cross-section geometry (110) on a JEM2100F (Jeol) electron microscope at an accelerating voltage of 200 kV. A standard method was used to prepare the sample, which involved thinning by precision grinding and spraying with argon ions during the final stage prior to perforation.

The obtained TEM image (Fig. 3) allowed identifying the regions of the MBLs, the InGaAs absorbing layer, and the InAl(Ga)As contact layers. In the image, the InP substrate is at the bottom and is separated from the metamorphic heterostructure by a bright white band. In the MBL, there was a large number of misfit dislocations in the form of horizontal dark bands and threading dislocations in the form of inclined dark lines. The dislocations were evenly distributed over the thickness of the MBL, and a rough estimate of the density of the extended defects in the buffer layer was ~ 1·10¹¹ cm⁻². Fig. 3

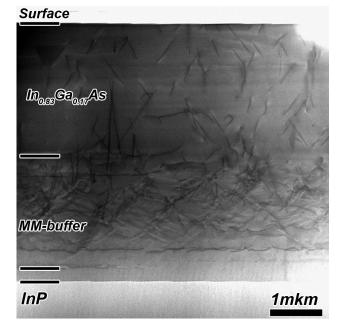


Fig. 3. STEM image of sample #1-LG

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shows that some dislocations were directly under the heterointerfaces of the InAs/InAlAs inserts, especially during the initial stages of the MBL growth. Also, threading dislocations were bending on various heterointerfaces, including in the DGSL region, however, their complete bending and annihilation occurred less often.

Earlier studies of test MBL heterostructures with a linear composition gradient revealed the existence of a dislocation-free region d_{free} [9]. Based on current images, it was impossible to judge whether the d_{free} region was present. What is more, the density of defects in the sample was much higher than the density of defects in the test MBL structure. Thus, according to the TEM images in the cross-sectional geometry, the density of the threading dislocations revealed in the functional layers of the $In_{0.83}Al_{0.17}As$ and $In_{0.83}Ga_{0.17}As$ heterostructures was $\sim 5\cdot 10^8$ cm⁻², while in the test structure, the density of dislocations in the In_{0.83}Ga_{0.17}As layer with a thickness of 500 nm was below the detection limit of the method, i.e. $< 1.10^6$ cm⁻² [9]. Perhaps this was due to the absence of the inverse step in the InAlAs composition gradient, which was introduced into the MBL design of the test structure. The absence of the inverse step could also affect the density of threading dislocations in the InGaAs absorbing layer.

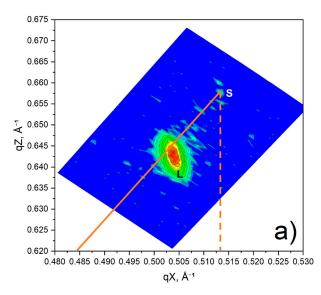
3.2. X-ray diffractometry and reciprocal space mapping of X-ray diffraction intensity

Information about the presence of elastic strain, misfit dislocations, and crystal lattice disorientation can be obtained from the X-ray diffraction reciprocal space mapping [13]. The X-ray diffraction reciprocal space map is a projection of the three-dimensional intensity distribution of the X-ray reflection intensity from the sample on the orthogonal axes of the reciprocal space qX < 110 > and qZ [001] which are parallel to the surfaces of the heterostructure and orthogonal to the surface of the heterostructure, respectively. The reciprocal space map is a set of contours of equal intensity, the greatest contribution to which is made by diffraction reflections from the selected family of crystalline planes. The X-ray scattering was recorded in the angular coordinates ω -2 θ , which were then converted into the reduced coordinates of the reciprocal space qX and qZ. X-ray diffraction spectra were recorded on a DRON-8 X-ray diffractometer (Burevestnik, Russia) with a Bartels monochromator and the radiation at the X-ray tube of $CuK\alpha 1 = 0.15406$ nm. The maps were converted and processed with the accompanying software package Reciprocal Space Mapping by *Burevestnik*.

Reciprocal space maps of X-ray diffraction intensity for reflections (224) for the samples were constructed and analyzed (Fig. 4). The reciprocal space mapping of X-ray diffraction intensity in the case of asymmetric reflection, for example, from a set of planes (224), allowed making judgements about the residual strain in the epitaxial layers. In Figures 4a and 4b, broad maxima of high intensity characterize diffraction on the functional layers of $In_{0.83}Al_{0.17}As/In_{0.83}Ga_{0.17}As$ of the 1-LG and 2-NLG samples, respectively. Lines drawn in the images in Fig. 4 characterize an entirely strainfree (solid line) and entirely pseudomorphic (dashed line) state of the crystal in relation to the substrate [14]. The line of the entire stain relaxation for reflection (224) always passes through the peak of the substrate, therefore, the position of the InGaAlAs maximum relative to this line can be used to conclude about the degree of strain relaxation in this layer. In both studied samples, the lines of strain relaxation passed near the reflection from the InGaAlAs layers, however, they did not cross its maximum. Thus, the active layers in both 1-LG sample and 2-NLG sample were partially strain-free.

The measurements of symmetric (004) reflections of the samples allowed making conclusions about the imperfection of the crystal lattice of the grown layers since the broadening of the intensity contours along the qX direction indicated dislocations or a mosaic structure, while the broadening along the qZ direction meant there was a fluctuation of the perpendicular component of the lattice parameter [15]. Let us assume that direction [110] of the X-ray beam incidence along the base cut of the substrate is an azimuthal angle of 0°, while direction [110], which is perpendicular to the base cut is an azimuth of 90°. X-ray scattering patterns from a set of reflecting planes (004) of the 1-LG and 2-NLG samples in the reciprocal space are shown in Fig. 5. The narrower and more intense

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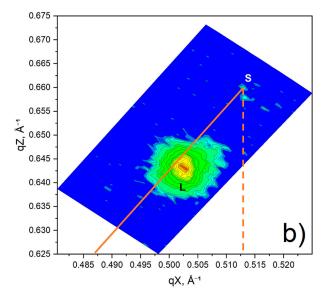


Fig. 4. Reciprocal space maps of asymmetric (224) reflection: a) sample #1-LG, δ) sample #2-NLG. Labels S and L stand for reflection peaks from InP substrate and In_{0.83}Ga(Al)_{0.17}As active layers. Orange lines represent fully relaxed (solid) and pseudomorphic (dashed) states

maximum at the top of all images corresponds to the InP substrate and is comparable to the angular coordinate $2\theta = 63.34^{\circ}$.

Let us consider Fig. 5a for a symmetrical reflection of the 1-LG sample with an azimuthal angle of 0° corresponding to the qX direction [110]. A continuous green region of medium intensity, elongated along the qZ axis, is characteristic of MBLs with a smooth composition gradient. The experimental maximum of the functional layers of ${\rm In_{0.83}Al_{0.17}As/In_{0.83}Ga_{0.17}As}$ (individual maxima of each layer are difficult to distinguish due to very close values of the lattice constant), located lower along the qZ axis, is shifted along the qXaxis relative to the maximum of the substrate, and, as compared to it, has a more pronounced ellipticity. The coordinate of the In_{0.83}Ga(Al)_{0.17}As peak along the qZ axis corresponds to an angle of $2\theta = 61.79^{\circ}$. This peak, as compared to the peak of the substrate, is much more stretched along both axes and its width in the qX direction exceeds many times the spread along qZ. Thus, we can assume that active layers of this sample have more local strain-free regions than regions with an elastically deformed lattice [15].

In the scattering pattern of the 2-NLG sample with similar conditions depicted in Fig. 5c, the maxima of diffraction intensity on the substrate and on the active layer are at the same positions along the qZ axis. However, in the case of the

2-NLG sample, the reciprocal space mapping did not reveal any significant reflection continuously stretched from the gradient buffer layer along the qZ axis. In addition, unlike the 1-LG sample, the maximum of the InGaAs layer was on the qX = 0line. What is more, it was elongated mainly along the qZ axis and was characterized by a much less expressed diffusion scattering along the qX axis than in Fig. 5a. As we noted above, the broadening of the peak along qZ may be due to a fluctuation of the lattice constant in a direction perpendicular to the scanning planes. Thus, a significant blurring of the maximum along the vertical axis may indicate the existence of local deformed regions of the crystal lattice and an inhomogeneous strain distribution over the crystal.

The half-width peaks of the intensity of the reflected X-ray emission along the ω axis in the angular coordinates of direct space ω -20 characterizes the defectiveness of the structure, which in the metamorphic heterostructure is mainly contributed by misfit dislocations of the screw type. The assessment of the dependence of the half-width intensity peaks relative to the ω axis on the normal lattice parameter $c = 2\lambda/\sin(\theta)$ [16] and the TEM images allowed characterizing the distribution of defects in the 1-LG sample. The results showed that the defectiveness of the linear MBL first increased with a thickness, was maximum in the middle and upper parts

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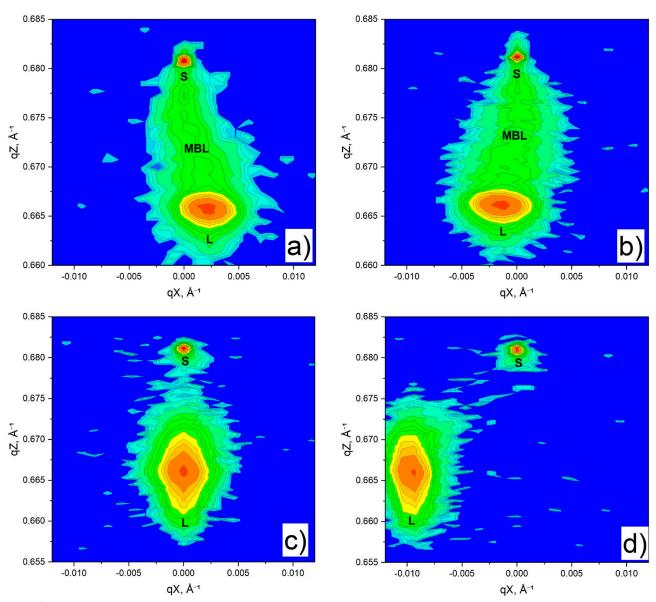


Fig. 5. Reciprocal space maps of symmetric (004) reflection of sample #1-LG (top) for azimuthal angles 0° (a) μ 90° (b) and sample #2-NLG (bottom) for 0° (c) and 90° (d). Labels S, L and MBL stand for reflection peaks from InP substrate, In_{0.83}Ga(Al)_{0.17}As active layers and In_xAl_{1-x}As metamorphic buffer layer

of the buffer layer, and then decreased slightly just before the formation of the DGSL and the absorbing layer.

It should be noted that when rotating the 1-LG sample by an azimuthal angle of 90° (Fig. 5b), the maximum lateral coordinate of $In_{0.83}Ga(Al)_{0.17}As$ changed its sign relative to the qX=0 axis, while the position of the peak along the qZ axis remained unchanged. However, similar to the case shown in Fig. 5a, the intensity spot from the linear MBL shifted uniformly towards the positive or negative qX, respectively. Fig. 5d shows images for the 2-NLG sample with an azimuthal angle of

90°. When rotating the sample with the root MBL by 90°, there was also a discrepancy between the maximum coordinates of the absorbing layer, however, in this case there was an abrupt change in the coordinate relative to qX = 0.

The mismatch between the maximum coordinates of the active layers and the substrate in the qX direction for the studied case of symmetrical reflection indicated the disorientation of the crystal lattice of the active layers relative to plane (001) [15]. The angle of the epitaxial layer disorientation relative to the substrate was characterized by inclination,

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rotation, or torsion. In lattice-mismatched heterostructures on substrates with orientation (001), disorientation could occur due to uneven distribution of dislocations over the surface relative to directions [110] and [110] [17]. This resulted in anisotropy of residual strain along these directions and could affect the angle of disorientation.

The reciprocal space mapping in the geometry of symmetrical reflections allowed estimating the inclination and rotation angles of the crystal lattice in the active layer relative to the plane of the substrate (001). For calculations, we used the relations given in [18, 19]:

$$tg(\alpha_{0^{\circ},90^{\circ}}) = \frac{\left| \Delta q X^{004} \right|_{0^{\circ},90^{\circ}}}{4 / a_{sub} - \left| \Delta q Z^{004} \right|_{0^{\circ},90^{\circ}}}$$

$$tg(\varphi) = \alpha_{0^{\circ}} / \alpha_{90^{\circ}}$$

where $\alpha_{\rm sub}$ = 5.8687 Å is the lattice constant of the substrate, qX^{004} is the difference in qX coordinates of the maxima of the layer and the substrate during symmetrical scanning (004), qZ^{004} is the difference in qZ coordinates of the maxima of the layer and the substrate during symmetrical scanning (004), $\alpha_{0^{\circ}}$ and $\alpha_{90^{\circ}}$ are the inclination angles of the crystal lattice at azimuthal angles of 0° and 90°, and ϕ is the azimuth of the zero inclination of the layer.

From Figures 5a and 5b, according to the given formulas, we obtained the inclination angles: $\alpha_{0^{\circ}} = 0.19^{\circ}$, $\alpha_{90^{\circ}} = 0.12^{\circ}$, and $\phi = 57.7^{\circ}$. By performing similar calculations and by using the data presented in Fig. 5c and 5d for the 2-NLG sample, we obtained $\alpha_{0^{\circ}}$ = 0°, $\alpha_{90^{\circ}}$ = 0.82°, and $\varphi = 0^{\circ}$. Therefore, in the case of root-like MBLs, the disorientation of the crystal lattice of the InGaAs layer was mainly affected by the inclination of the layer relative to plane (001). What is more, there was no torsion of the lattice, which can also be indirectly confirmed by scanning at coordinates ω - φ . According to [17], large values of angles α correlate with low efficiency of elastic strain relaxation in the structure through the formation of dislocations. Therefore, the disorientation of the crystal lattice is a secondary mechanism for relieving residual strain. In the case of the 2-NLG sample, this may mean that during the early stages of the buffer layer growth, the strain relaxation mechanism was mainly due to the formation of dislocations, while further it was a tilt of the lattice that was responsible for the release of excessive strain.

4. Conclusions

In our study, for the growth of active layers of In_{0.83}Ga(Al)_{0.17}As on InP substrates, we proposed two alternative designs of the metamorphic buffer layer: one following a linear law of the composition variation and the other following a nonlinear law, which in the studied case had a square root-like dependence. To determine the effectiveness of these metamorphic buffer layers, we used transmission electron microscopy and the reciprocal space mapping of X-ray diffraction intensity to study the crystalline quality of the bulk $In_{0.83}Ga(Al)_{0.17}As$ layers grown on the buffer layers. The reciprocal space mapping for asymmetric reflections from a set of planes (224) showed that the active layers were partially strain-free in heterostructures with both linear and nonlinear buffer layer. They also revealed the formation of misfit dislocations. The reciprocal space mapping in the geometry of symmetrical reflections (004) allowed establishing that in both samples elastic strain was relieved through the disorientation of the crystal lattice relative to the plane of the substrate. The density of threading dislocations in the active layers of ${\rm In}_{\rm 0.83}{\rm Ga(Al)}_{\rm 0.17}{\rm As}$ in the heterostructure with a linear metamorphic buffer layer calculated from the images of transmission electron microscopy was ~ 5·108 cm⁻². Presumably, in order to reduce it and to obtain a dislocationfree region in the heterostructure, the growth of the linear metamorphic buffer layer must be completed with an inverse step with an increase in the mole fraction of indium in the composition of the buffer layer relative to the composition of the active layers.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Study of hyperfine interactions in spinel cobalt ferrite $CoFe_2O_4$ doped with Hf, Lu, and Yb using Mössbauer spectroscopy and perturbed γ - γ angular correlation

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Abstract

We studied hyperfine interactions (HFI) in spinel cobalt ferrite (ferrospinel) $CoFe_2O_4$ doped with Lu, Yb, and Hf (1 wt. %) by Mössbauer spectroscopy (MS) on ⁵⁷Fe nuclei. The interactions indicate the presence ($CoFe_2O_4$:Lu – 11 %, $CoFe_2O_4$:Yb – 23.4%) and the absence ($CoFe_2O_4$:Hf) of additional phases. The study revealed a significant change in the HFI parameters on ⁵⁷Fe nuclei in the octahedral sites in ferrospinels doped with Hf, Lu, and Yb. However, the relative influence of the impurity on ⁵⁷Fe nuclei in the tetrahedral sites was insignificant.

The parameters of hyperfine fields on 172 Yb nuclei in cobalt ferrite were obtained by the method of perturbed γ - γ angular correlations (PAC). The 172 Yb ions were introduced into the sample using two methods: by adding 172 Hf and 172 Lu isotopes. A significant difference in the HFI parameters for these two cases was revealed. The local environment of 172 Yb ions appears to be different in the two variants of isotope introduction into the sample (Hf or Lu). The difference in the HFI parameters persisted in the temperature range of 300 - 1000 K.

It was determined that the different effects of Hf and Lu on the parameters (electric field gradient, magnetic field, and isomer shift) of the HFI in the sample are revealed by both MS and PAC methods, irrespective of the amount of the dopant. According to the MS data, Hf and Lu do not lead to significant changes in the HFI parameters in the tetrahedral sites ("Sextet 1"), but have a significantly different effect on the same parameters in the octahedral sites ("Sextet 2"). A similar pattern was observed using the PAC method: replacing Hf with Lu did not reveal any changes in the HFI parameters in one of the sites (the octahedral site), but indicated a significant change in the other site.

Keywords: Mössbauer spectroscopy, Perturbed correlations, Spinels, Oxides, Ferrites

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1. Introduction

Ferrites are mixed metal oxides with the general formula MFe₂O₄, where the metal (M) may be represented by Fe, Ni, Co, Zn, etc. They have a number of properties (high resistivity, high magnetic permeability, and high penetration depth of microwave field) that contribute to their application in various fields [1–9]. Cobalt ferrite (CoFe₂O₄) is one of the most widely used magnetic materials. In modern technologies, ferrites are often doped, because even small (≈ 1 %) amounts of additives significantly change their properties. Currently, rare earth elements (RE) are considered as dopants [6-15]. In the case of modification with RE3+ ions, taking into account their typical oxidation state of +3, we can expect them to replace Fe^{3+} [11–15]. On the other hand, considering that the ionic radius of the RE is larger than the ionic radius of Fe³⁺, such substitution is difficult, and an additional phase may be formed [14]. Ferromagnetic spinel semiconductors CdCr₂Se₄ and CdCr₂S₄ were studied by the PAC method at the ISOLDE on-line separator (CERN) [16]. But in that work, ferrites were not studied, and besides the introduced isotopes (111In, 111mCd, 111Ag, 117Cd, and 199mHg) were not rare earth elements. In work [17], nonrare earth impurities of CoFe₂O₄ were studied by Mössbauer spectroscopy.

The aim of this work was to study the hyperfine interactions in spinel cobalt ferrite (ferrospinel) CoFe₂O₄ doped with Lu, Yb, and Hf by Mössbauer spectroscopy (MS) and perturbed angular correlation (PAC) methods.

2. Experimental

2.1. Synthesis of samples

For the study, samples of cobalt ferrite CoFe₂O₄ were obtained by adding dopants during the synthesis. In the first case (the MS method), natural Yb, Lu, and Hf were added in the amount

of 1 wt. % of the total weight of the sample. For the PAC measurements, CoFe₂O₄ was produced with the addition of radionuclides during the synthesis.

Ferrites were synthesized from mixed aqueous solutions of Fe and Co salts with a molar ratio of 2:1 by precipitation with 1 M NaOH solution to the pH of 11-12. To modify the samples with stable Yb, Lu, and Hf or radionuclides, solutions of Yb, Lu, or Hf nitrates or solutions of ¹⁷²Lu or ¹⁷²Hf radionuclides were added to the initial aqueous solutions of Fe(III) and M(II) (M = Co) salts. The procedure was described in detail in [13, 14, 15]. The preparation of radionuclides was described in [18]. The final washing of the precipitate was carried out with a 50/50 (v/v) water/acetone mixture. The washed samples were dried at 80 °C overnight and then heat treated at 750 °C in air for 5 hours. The obtained mixed metal oxides based on CoFe₂O₄ were labelled as CFO.

2.2. Mössbauer spectroscopy

The Mössbauer spectra were measured on a MS-1104Em spectrometer [20] in the sample mass range from 1 to 5 mg. The spectrometer was operated in transmission geometry at room temperature. The 57Co radiation source was modulated in a mode in which the dependence of the Doppler velocity over time has a triangular shape. A 57Co source in a Cr matrix with an activity of 30 mCi produced by Ritverc JSC was used as a resonant source of γ -quanta. A scintillation detector based on a NaI scintillator was used. It was calibrated against the metallic α -Fe. The speed was 12.11 mm/s, the triangular velocity profile of the absorber was used to record the Mössbauer spectra at forward and reverse movement. The Mössbauer spectra were approximated by Lorentzian lines according to the χ^2 criterion.

2.3. Perturbed γ - γ angular correlation method

The PAC method used in studies [20, 21, 22] is based on the introduction of a radioactive isotope into the sample, the decay of which is accompanied by the emission of cascade γ -rays. The advantage of the PAC method is the very low concentration of probe nuclei, so it does not change the properties of the studied sample. We used the isotopes ¹⁷²Lu(¹⁷²Yb) and ¹⁷²Hf(¹⁷²Lu(¹⁷²Yb)) for the PAC measurements, the daughter nuclides are provided in brackets.

The PAC measurements were performed using a 4-detector (BaF_2 crystals) spectrometer [22]. The detectors were positioned in the same plane at 90° to each other. The studied sample was placed in the center between them. The anisotropy of angular correlation is determined by the formula:

$$R(t) = A_2 G_2(t) Q_2 = 2 \frac{S(180^\circ, t) - S(90^\circ, t)}{S(180^\circ, t) + 2S(90^\circ, t)}, \quad (1),$$

where S (90°, t) are the gamma spectra of delayed coincidences when the detectors were placed at 90° (eight possible combinations of two detectors); S (180°, t) are the spectra of delayed coincidences when the detectors were placed at 180° (four possible combinations of two detectors); and Q_2 is a geometric factor (corresponding to the sizes and types of detectors and sources). The maximum time interval for obtaining delayed coincidence spectra was 800 ns. The timing reso-

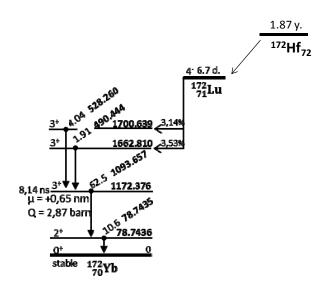


Fig. 1. Schemes of the decay of ¹⁷²Hf and ¹⁷²Lu in the target region for the PAC measurements on ¹⁷²Yb [24]

lution (60 Co, 1173-1332 keV) for a BaF $_2$ scintillator was 400 ps.

The decay scheme of 172 Lu and the used cascades of 172 Yb γ -quanta are presented in Fig. 1. Since 172 Hf decays into 172 Lu [23], either 172 Lu or 172 Hf were incorporated into the sample for the PAC measurements with the isotope 172 Yb.

3. Results and discussion

3.1. Studies of ferrites with the addition of 1% stable Hf, Yb, and Lu

Powder samples of CFO spinel ($CoFe_2O_4$) doped with Hf, Lu, and Yb (1 wt. %) were studied. Hereinafter they are denoted as $CoFe_2O_4$:Hf, $CoFe_2O_4$:Yb, and $CoFe_2O_4$:Lu. Fig. 2 shows the ⁵⁷Fe Mössbauer spectra for these samples. The spectra were measured at room temperature.

In the crystal lattice of a spinel ferrite, iron cations occupying the B sites (octahedral) have oxidation degrees +2 and +3, while in the A sites (tetrahedral) they have only +3. Each site has its own sextet in the spectrum ('Sextet 1' – (CoFe₂O₄) A and 'Sextet 2' - [CoFe₂O₄]B), which carries information about magnetic dipole and electric quadrupole interactions. Both sextets are indicated by the smooth fit lines against the spectrum background. Also, the spectra clearly show the manifestation of the impurity doublet (additional phase) in the cases of Lu and Yb. In the case of Hf, there is no impurity doublet (phase). All parameters for processing Mössbauer spectra are presented in Table 1. X-ray diffraction studies of powdered CoFe₂O₄:Hf, CoFe₂O₄:Yb, and CoFe₂O₄:Lu samples were carried out using a DRON-3 diffractometer at the KaMo line in the reflection mode [24]. The X-ray diffraction of our samples showed the presence of an additional phase only in CoFe₂O₄:Yb. In an earlier study [15], no second phases were also observed in the X-ray diffraction patterns of CoFe,O, samples doped with Lu and Hf.

The analysis of the parameters of the Mössbauer spectra showed that the $CoFe_2O_4$:Hf sample contains 100 % standard spinel sextets ($CoFe_2O_4$)A and [$CoFe_2O_4$]B, the $CoFe_2O_4$:Lu sample contains 11 % of the impurity CoFe phase besides the standard sextets, and the $CoFe_2O_4$:Yb sample contains 23.4 % of the impurity CoFe phase.

The data for the ⁵⁷Fe isomer shifts (Fig. 3) in CoFe₂O₄ (1 wt. %) Hf, Lu, and Yb indicate a strong influence of the impurity on ⁵⁷Fe in the B sites and

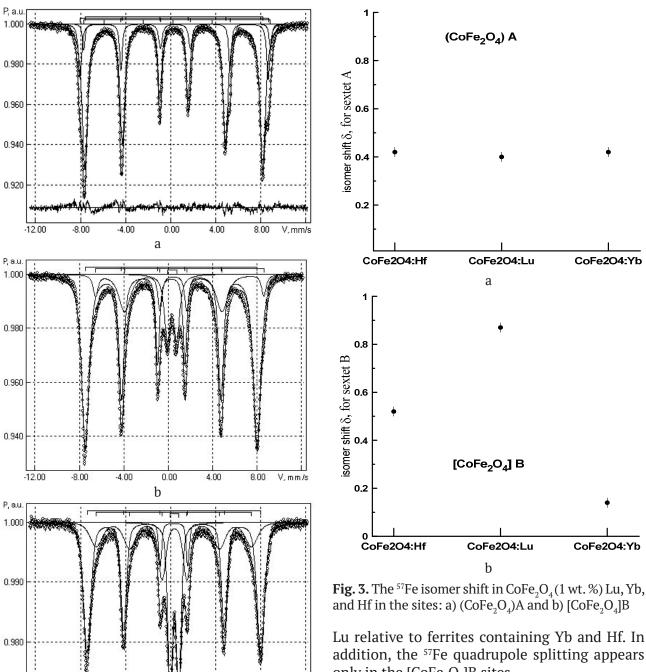


Fig. 2. Mössbauer spectra for ⁵⁷Fe in samples: a) CoFe₂O₄:Hf; b) CoFe₂O₄:Lu, and c) CoFe₂O₄:Yb, measured at 298 K

0.00

4.00

8.00

a negligible influence in the A sites.

-4.00

-12.00

-8.00

Studying the 57Fe quadrupole state in CoFe₂O₄ (1 wt. %) Hf, Lu, and Yb ferrites (Fig. 4) also provided important results. We observed a large ⁵⁷Fe quadrupole splitting in CoFe₂O₄ (1 wt. %)

Lu relative to ferrites containing Yb and Hf. In

addition, the 57Fe quadrupole splitting appears only in the [CoFe₂O₄]B sites.

The results for the 57Fe hyperfine magnetic field in CoFe₂O₄ (1 wt. %) Hf, Lu, and Yb are shown in Fig. 5a, b. We observed a significant change in the ⁵⁷Fe hyperfine field in the [CoFe₂O₄] B site depending on the impurity element. The influence of the impurity element on ⁵⁷Fe in the [CoFe₂O₄]A sites was negligible.

3.2. Studies of CoFe₂O₄ ferrites by the PAC method

For the PAC measurements on ¹⁷²Yb nuclei, we can add either 172Lu or 172Hf radioactive

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Table 1. Mössbauer parameters of $CoFe_2O_4$ samples. δ is the isomer shift, ΔE_Q^a is the quadrupole splitting, G is the line width, H is the Fe magnetic field, and A is the area under the spectrum

Sample	Sextet in the spectrum	δ (mm/s)	$\Delta E_{\rm Q}^{\rm a} ({\rm mm/s})$	G (mm/s)	H (kOe)	A (%)	Component
CoFe ₂ O ₄	«Sextet_1»	0.42	-0.00	0.56	490	77.40	$(CoFe_2O_4)A$
+1 % Hf	«Sextet_2»	0.52	-0.06	0.44	518	22.60	[CoFe ₂ O ₄]B
CoFe ₂ O ₄	«Sextet_1»	0.40	-0.00	0.80	480	68.50	$(CoFe_2O_4)A$
+1 % Lu	«Sextet_2»	0.87	0.57	0.50	470	20.46	$[CoFe_2O_4]B$
	«Doublet_1»	0.47	0.84	0.63		11.04	CoFe
CoFe ₂ O ₄	«Sextet_1»	0.42	-0.00	0.79	478	48.83	$(CoFe_2O_4)A$
+1 % Yb	«Sextet_2»	0.14	-0.06	1.29	427	27.87	[CoFe ₂ O ₄]B
	«Doublet_1»	0.14	0.74	0.51		23.39	CoFe

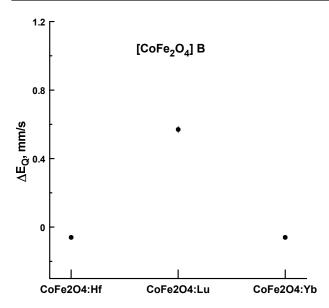
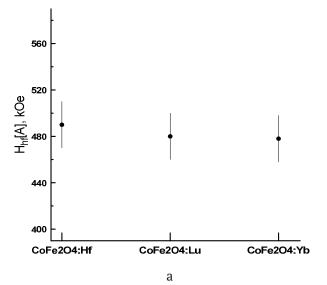


Fig. 4. The ⁵⁷Fe quadrupole splitting in CoFe₂O₄ (1 wt. %) Hf, Lu, and Yb ferrites

isotope to the sample. It should be noted that the radioactive isotopes were added without a carrier. An important assumption, apparently quite obvious, is that the position of the probe nucleus in the sample matrix coincides with the position of the parent nucleus.

The results at room temperature (considerably below the Curie temperature) showed that the CoFe $_2$ O $_4$ ferrite (CFO) with incorporated 172 Hf (Fig. 6, above) provided two states of 172 Yb, characterized by two magnetic fields B $_{\rm hf1}$ = 14(1) T and B $_{\rm hf2}$ = 17(1) T with a population ratio of $^{\sim}$ 2:1. In the case of incorporated 172 Lu (Fig. 6, below), the processing of the spectrum indicated the presence of a single magnetic field B $_{\rm hf}$ = 18(1) T. This significant difference can be explained by the assumption that at room temperature 172 Lu is introduced preferentially into only one type of crystal lattice sites.



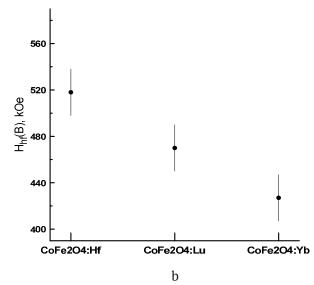


Fig. 5. The ⁵⁷Fe hyperfine magnetic field in $CoFe_2O_4$ (1 wt. %) Lu, Yb, and Hf ferrites in the sites: a) $(CoFe_2O_4)A$ and b) $[CoFe_2O_4]B$

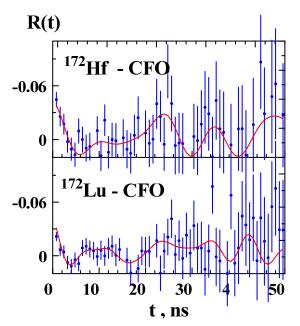


Fig. 6. The ¹⁷²Yb PAC spectra in CoFe₂O₄ ferrite measured at 298 K, after the introduction of ¹⁷²Hf (above) and ¹⁷²Lu (below) into the sample

It should be noted that below the Curie temperature, the hyperfine field is a mix of magnetic dipole and electric quadrupole fields. To reveal more clearly the influence of dopant ions (Hf and Lu) on the electric quadrupole interaction (directly related to the position of the dopant ion in the crystal lattice), we carried out measurements above the Curie temperature (1000 K). Table 2 shows the obtained quadrupole frequency, electric field gradient, and relative site occupancy parameters for ¹⁷²Yb in CoFe₂O₄(CFO), after introducing ¹⁷²Lu or ¹⁷²Hf into the sample.

The observed difference in the parameters can be explained by the assumption that the ³⁺Lu and ⁴⁺Hf ions are distributed differently when introduced into two non-equivalent lattice sites: in addition to the different charges of these ions, the lutetium ion has a larger ionic radius.

Basically, the post-effects of nuclear decay may influence the local environment of $^{172}{\rm Yb}$

in CFOs. However, in crystalline phases, these effects usually have no significant influence on the HFI parameters.

4. Conclusions

Using the Mössbauer spectroscopy on ⁵⁷Fe nuclei, we studied the hyperfine interactions in CoFe₂O₄ ferrites doped with Hf, Lu, and Yb (1 wt. %). It was shown that doping with Lu and Yb leads to the formation of additional nonmagnetic phases (Lu -11% and Yb -23.4%). In the ferrite doped with hafnium Hf, no additional phase was observed. A significant change in the HFI parameters was revealed for ⁵⁷Fe in the octahedral sites of CoFe₂O₄ ferrites doped with Hf, Yb, or Lu, respectively. However, the relative influence of the impurity on 57Fe nuclei in the tetrahedral sites was insignificant. The formation of a new phase in the case of Yb and Lu can be attributed to the fact that the ionic radii of Yb and Lu are about 14-13 % larger than the ionic radius of Hf.

The parameters of hyperfine interactions on $^{172}{\rm Yb}$ nuclei in ${\rm CoFe_2O_4}$ ferrites were obtained by the PAC method. The $^{172}{\rm Yb}$ ions were introduced (10-7 - 10-8 wt. %) into the samples in two ways: via $^{172}{\rm Hf}$ ($^{172}{\rm Hf}$ \rightarrow $^{172}{\rm Lu}$ \rightarrow $^{172}{\rm Yb}$ chain) or via the parent $^{172}{\rm Lu}$. A significant difference in the HFI parameters for these two cases was revealed. The difference in the HFI parameters persisted in the temperature range of 300–1000 K. The observed difference is due to the different distribution of $^{3+}{\rm Lu}$ and $^{4+}{\rm Hf}$ over the lattice sites due to different charges and ionic radii of these ions.

Notably, the different effects of Hf and Lu on the parameters (electric field gradient, magnetic field, and isomer shift) of the HFI in the samples were revealed by both MS and PAC methods, irrespective of the amount of the dopant. Specifically, according to MS, Hf, and Lu did not cause significant changes in the HFI parameters in the tetrahedral sites ("Sextet 1"),

Table 2. The parameters of the hyperfine interactions (above the Curie temperature) for 172 Yb upon the introduction of 172 Lu or 172 Hf into CFOs: quadrupole frequency ω_Q , electric field gradient Vzz, and relative site occupation f

	ω_{Q1} , Мрад/с	V zz, 10^{21} B/m ²	$f(\omega_1), \%$	ω _{Q2} , Мрад/с	V zz, 10^{21} B/m 2	$f(\omega_2)$, %
172 Hf(172 Lu(172 Yb))	220(15)	10.12(69)	65(8)	60(8)	2.76(37)	35(7)
¹⁷² Lu(¹⁷² Yb)	231(5)	10.63(56)	55(7)	108(9)	4.97(41)	45(8)

but had significantly different effects on the same parameters in the octahedral sites ("Sextet 2"). A similar situation was observed using the PAC method: substitution of Hf by Lu did not reveal a change in the HFI parameters in one of the sites (according to [15], it is the octahedral site), but indicated a significant difference (see Table 2) in the other site.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Chemical vapor deposition of Tm₃Fe₅O₁₂ epitaxial films, investigation of their structure and properties in the terahertz range

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Abstract

In this study, for the search and development of new spintronic materials, thin films of $\mathrm{Tm}_3\mathrm{Fe}_5\mathrm{O}_{12}$ iron garnet were obtained by the metalorganic chemical vapor deposition (MOCVD) on single-crystal $\mathrm{Gd}_3\mathrm{Ga}_5\mathrm{O}_{12}(111)$ – GGG and $\mathrm{Y}_3\mathrm{Al}_5\mathrm{O}_{12}(111)$ – YAG substrates. The $\mathrm{Tm}_3\mathrm{Fe}_5\mathrm{O}_{12}$ films were investigated using X-ray diffraction, Energy dispersive X-Ray microanalysis, Raman spectroscopy and terahertz (THz) pulsed spectroscopy.

The epitaxial nature of films deposited on substrates of both types demonstrated. It was found that the growth of garnet film under the high-temperature vacuum conditions of MOCVD on a GGG substrate is complicated by the evaporation of gallium oxide, which causes the introduction of iron oxide into the surface layer of the substrate, enrichment of the adjacent layer of the film with thulium oxide and the formation of non-stoichiometric garnet with antisite defects.

It was concluded that YAG substrates are more promising, since the heteroepitaxy of iron garnets on them does not have such complications.

Keywords: Thin films, Iron garnets, MOCVD, Structure, Antisite defects, Raman spectroscopy, Terahertz spectroscopy

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1. Introduction

The successes achieved in the last decade in the development of new areas of electronics – spintronics and magnonics are based on the development and application of new materials, including those combining several functional properties [1]. One of the main trends in modern engineering is the miniaturization of electronic devices, so research and development of multifunctional materials in the form of thin films is of particular interest. This direction of development is especially relevant for spintronics and magnonics, which use ferrimagnets with a garnet structure in the form of ultra-thin (from units to tens of nanometers) epitaxial films as the main materials [1]. To obtain such thin films, the liquid-phase epitaxy method, which has long been the most popular method for synthesizing garnets (the thickness of the films obtained in this case is a few micrometers [2, 3]), cannot be used; therefore, new technological approaches to obtaining thin films of iron garnets are required. One of the promising synthesis methods is the metalorganic chemical vapor deposition (MOCVD) using, which has become the main method in the synthesis of epitaxial heterostructures of semiconductors [4]. Using b-diketonates and alcoholates as volatile substances, MOCVD can also be quite successfully applied to the epitaxy of functional materials based on simple and complex oxides [5].

The structural class of garnets belongs to the space group *Ia3d*. The cubic body-centered unit cell of garnet contains 8 identical octants corresponding to the formula composition {RE}₃[Fe]₂(Fe)₃O₁₂, where the curly brackets { } denote the dodecahedral positions of the rare earth element (RE), and the brackets []- and ()-correspond to the octahedral and tetrahedral positions occupied by Fe³⁺ ions. Thus, the features of iron-based garnets are: a multi-sublattice magnetic structure, have wide possibilities for varying the elemental composition and, as a consequence, a variety of their magnetic properties [6].

Iron garnets of rare earth elements (REE) have high resistance and, accordingly, high transparency in the visible and near IR ranges, which gives them an advantage over metallic ferromagnets (Fe, Ni, etc.), since it allows the use

of the pump-probe technique in the study of ultrafast magnetization dynamics [1,7]. Spin pumping is also a way to excite magnons in magnetic systems. Recently, spin wave propagation in a lutetium iron garnet/heavy metal ($\text{Lu}_3\text{Fe}_5\text{O}_{12}/\text{Pt}$) is demonstrated in [8]. The results of such studies, along with great fundamental interest, open up prospects for the creation of ultra-high-speed devices for recording and processing information using thin ferrimagnetic films [9].

Spin-flip transitions of some rare earth iron garnets occur in the frequency range from 0.1 THz to 10 THz. Therefore, terahertz radiation can potentially affect the spin state of such systems and can be used to obtain important results in the study of magnetic dynamics under the influence of laser radiation [10], however, research in this direction is in the initial phase of its development. Systematic studies of the spectral properties of iron garnet films in the THz range and their changes in a number of rare earth elements are of great interest.

Thus, the aim of this study was the MOCVD synthesis of thin films of ${\rm Tm_3Fe_5O_{12}}$ iron garnet, comparative analysis of the results of epitaxy on isomorphic ${\rm Gd_3Ga_5O_{12}}$ and ${\rm Y_3Al_5O_{12}}$ (111) substrates, investigation of the structure of the obtained films by X-ray diffraction, Raman spectroscopy, and also study of their spectral characteristics by time-resolved pulsed THz spectroscopy.

2. Experimental

Thin $Tm_{_{3}}Fe_{_{5}}O_{_{12}}$ films were synthesized on single-crystal Ga₃Ga₅O₁₂ (GGG) and Y₃Al₅O₁₂ (YAG) substrates with the (111) orientation by metal-organic chemical vapor deposition (MOCVD). Dipivaloylmethanates Tm(thd)₃ and $Fe(thd)_{3}$ (thd – 2,2,6,6-tetramethylheptane-3,5dionate anion) were used as volatile compounds (precursors). In our previous study, we discussed in detail a new principle of dosing precursor vapor into the reactor, which ensures smooth, well-controlled and reproducible "feeding" of the growing film [11]. The scheme of the modernized setup using this principle and applied in this study is shown in Fig. 1. The system operates by passing a cotton thread impregnated with an metalorganic precursor solution through two separate vacuum zones - a low-temperature solvent evaporation

zone and a higher-temperature zone where precursors sublimate. The solvent vapors (toluene was used) are condensed in a nitrogen trap, and the precursor vapors are transferred by a carrier gas flow (Ar) into a vertical hot-wall reactor, where their oxidative thermolysis occurs with the formation of an oxide film on a heated substrate. The substrate holder with the substrate fixed to it rotates around a vertical axis in order to increase the lateral homogeneity of the films by thickness. The vacuum in the system is created by a fore vacuum pump equipped with a valve for regulating the overall pressure in the film deposition zone. The temperature of the hot lines through which the precursor vapors are supplied and the temperature of the substrate holder were 190 and 910 °C respectively; the total pressure was 10 mbar, the partial pressure of oxygen was 1 mbar, the film growth rate was 3 nm/min. After film deposition, the setup was filled with oxygen to atmospheric pressure and oxidative annealing was carried out at the spraying temperature for 20 min, after which the setup was cooled. The thickness of the films was 900 nm.

The Fe/Tm ratio in the obtained films was determined by Energy dispersive X-Ray

microanalysis (EDX) data using a Zeiss EVO 50 SEM scanning electron microscope with an e2v Sirius SD EDX analyzer. The phase composition and orientation of the obtained films were determined based on X-ray θ -2 θ scanning data using a Rigaku SmartLab diffractometer (Cu $K\alpha$, graphite monochromator). Measurement in the θ -2 θ scanning mode was carried out in the angle range of 5–80° with a step of 0.02°, the signal accumulation time was 1 s. Phase analysis was performed using the ICDD PDF database. For the determination of the in-plane orientation of the films and confirm epitaxial growth, X-ray ϕ -scanning was performed.

Raman spectra at wavelengths of 532 and 785 nm were obtained using a HORIBA Scientific Raman spectrometer and microscope. The measurement range was from 80 to 2000 cm⁻¹. Raman spectroscopy was performed with a 100x objective at an incident radiation intensity of 140 W/cm² for λ = 532 nm and 3.6·10³ W/cm² for λ = 785 nm. Three independent measurements were performed, each of which was averaged over 30 scans with an accumulation time of 0.5 s at each point. The background component of the spectrum was subtracted using the Voight

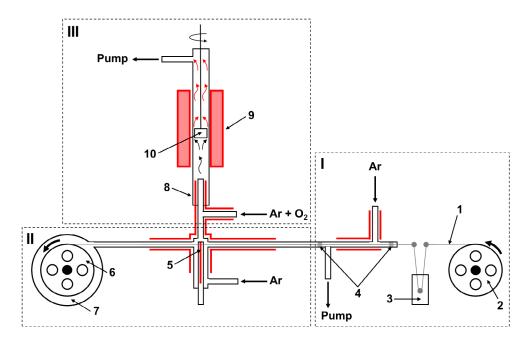


Fig. 1. Schematic diagram of MOCVD setup with thread-solution feeder. **I** – Thread feeding and solvent evaporation unit: 1 – cotton thread, 2 – feeding reel, 3 – reservoir for precursor solution, 4 – dies. **II** – Precursor evaporation unit: 5 – precursor evaporator, 6 – receiving reel, 7 – vacuum cover of the receiving reel. **III** – Film deposition unit: 8 – quartz reactor, 9 – reactor furnace, 10 – rotating substrate holder. Heated lines are marked in red

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function approximation of the obtained Raman spectra.

Terahertz spectroscopy in both transmission and reflection geometry was performed using a commercial TeraSmart THz spectrometer (Menlo Systems GmbH, Germany). Linear p-polarized THz pulses are generated in a fiber-coupled photoconductive antenna (Fe:InGaAs/INASAS) and measured by two photoconductive detectors (LT InGaAs/InAlAs), which simultaneously measure the transmitted and reflected THz beams. The spectrometer has a dynamic range of 95 dB, a spectral range of 0.2-3.6 THz and a scan window of 110 ps, providing a frequency resolution of 7.3 GHz. The spectral range for measurements is limited in the low-frequency region by the sample aperture (4 mm), and in the high-frequency region (above 3.6 THz) by the detectors' absorption lines. The THz radiation was focused onto the sample by two parabolic mirrors, and the THz beam size on the sample was about 500 μ m at the $1/e^2$ level. In this study, the incidence angle was 35°. To reduce the effect of water vapor absorption, the experiments were conducted using a dry air system that maintained a relative humidity of 20% in the chamber. In addition, to reduce systematic error, 3 independent measurements of 1000 pulses each were performed for each sample, the results of which were averaged.

The method for obtaining the absorption coefficient and refractive index for the transmission configuration is described in [12,13]. Determination of the dependence of the amplitude of the transmitted field on the frequency (w) is possible using the Fourier transforms (FFT) of the time profile of the THz pulse E(t):

$$E_{\text{sample}}(\omega) = FFT(E(t)). \tag{1}$$

The transmittance coefficient $T_w(\omega)$ of the sample is calculated as the ratio of the amplitude of the field passed through the sample to the amplitude of the field without the sample (air):

$$T_{w}(\omega) = \frac{\text{substrate}(\omega)}{E_{\text{air}}(\omega)}$$
 (2)

From here the absorption index of the sample is calculated as:

$$\alpha(\omega) = \frac{-\ln(T_w(\omega)) + \ln(1 - R^2)}{d},$$
 (3)

$$n(\omega) = n_{\text{aver}} + \arg(T_w(\omega)) \frac{c}{\omega d}, \tag{4}$$

where d – sample thickness, $R = \frac{n_{\text{aver}} - 1}{n_{\text{aver}} + 1}$ – reflec-

tion coefficient, $n_{\text{aver}} = 1 + \Delta t \cdot \frac{c}{d}$ – average refrac-

tive index, Δt – pulse delay when passing through the sample, c – the speed of light.

3. Results and discussion

For deposition of ${\rm Tm_3Fe_5O_{12}}$ films two substrates with a garnet structure were selected – GGG(111) and YAG(111) with very different unit cell parameters (Fig. 2), which leads to different values of the unit cell parameter mismatch (e) at the film/substrate interface. Values e, calculated using the formula:

$$\varepsilon = \frac{a_{\text{substrate}} - a_{\text{film}}}{a_{\text{substrate}}} \cdot 100 \%, \tag{5}$$

were 0.4% and -2.8% for films on GGG(111) and YAG(111), respectively. As can be seen from the calculated values, during epitaxy on YAG, the substrate compresses the film in-plane, and in the case of GGG, on the contrary, the substrate stretches the film in-plane. Thus, during the growth of ${\rm Tm_3Fe_5O_{12}}$ films on these substrates, significant elastic stresses can be expected in the films, which usually lead to changes in many physical properties, in particular magnetic ones.

The $\theta/2\theta$ diffraction patterns of $Tm_3Fe_5O_{12}$ films, obtained by simultaneous deposition on GGG(111) and YAG(111) substrates are shown

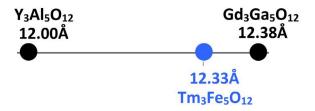


Fig. 2. Schematic illustration of the difference in the unit cell parameters of $Gd_3Ga_5O_{12}(111)$ and $Y_3A-l_5O_{12}(111)$ garnet substrates and $Tm_3Fe_5O_{12}$ film

in Fig. 3a. It is evident that in both diffraction patterns, in addition to the substrate peaks, there were peaks from the Tm₃Fe₅O₁₂ film (444), however, on the GGG(111) substrate there were also additional peaks, designated 2 and 3 (Fig. 3a, b). This result is identical to recently observed in our study of the structure of lutetium iron garnet $\text{Lu}_3\text{Fe}_5\text{O}_{12}$ films on GGG substrates [14]. Under MOCVD synthesis conditions ($T_{\text{deposition}}$ = 890-950 °C and low partial pressure of oxygen ~1 mbar) surface chemical instability of the Gd₃Ga₅O₁₂ substrate was observed due to the volatility of gallium oxide Ga, O. This leads to the formation of a surface layer of Gd_x(Fe_yGa_{x-y})O₁₂ in the substrate, which occurred as a result of filling vacancies formed during partial loss of gallium occupied by iron ions, were formed primarily in octahedral positions. The diffraction peak 2 in Fig. 3 indicates the formation of the epitaxial layer of $Gd_3(Fe_xGa_{5-x})O_{12}$. The calculated unit cell parameter for the $Gd_3(Fe_xGa_{5-x})O_{12}$ layer was 12.46 Å, which was close to the unit cell parameter for $Gd_{3}Fe_{5}O_{12} = 12.50 \text{ Å}.$

Diffraction peak **3** (Fig. 3a, b) we interpret as belonging to the intermediate Tm-excessive layer of garnet of composition $Tm_3(Tm_xFe_{5-x})O_{12}$, similar to $Lu_3(Lu_xFe_{5-x})O_{12}$, observed in [14] and formed as a result of partial occupation of octahedral positions of garnet by Lu^{+3} ions and a decrease

in their occupation by Fe⁺³ ions. Since the ionic radius of Tm⁺³ (coordination number = 6) is higher than that of Fe⁺³ (0.88 Å and 0.65 Å, respectively [15]), then such a substitution leads to a strong increase in the unit cell parameter of the Tm⁺³ compositions as the x value increases, which is manifested in a shift in the X-ray reflection of garnet (444) to the region of smaller angles. The possibility of the location of REE ions not only in dodecahedral but also in octahedral positions was pointed out by Geller in his classic study in 1967 on the crystallography of garnets [6]. Also the theoretical study[16], which examined these possibilities in detail and proposes an algorithm for calculating the degree of occupation of oxygen octahedra in garnet single crystals by REE ions, as well as the study [17], in which the formation of such antisite defects in a number of garnets, studied by the EXAFS method, was considered as the cause of a decrease in the cubic symmetry of garnets should be mentioned. At the same time, systematic and convincing experimental evidence of such a redistribution of REE ions by positions, as well as reliably established information about the region of non-stoichiometry of garnets in the systems "REE oxide – A_2O_3 (A = Al, Ga, Fe)" could not be found in the literature. In addition, it should be noted that the possibility of formation of antisite defects $[RE_{_{Fe}}]^{^{3+}}$ in heteroepitaxial films and autonomous phases (powders, single crystals)

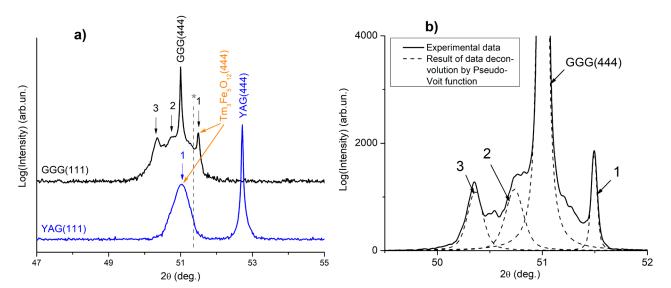


Fig. 3. (a) X-ray diffractograms of $Tm_3Fe_5O_{12}$ films deposited on GGG(111) and YAG(111) substrates. The dashed line indicates the (444) peak position of bulk $Tm_3Fe_5O_{12}$. (b) Magnified fragment of the diffractogram of $Tm_3Fe_5O_{12}$ film deposited on GGG(111) with description of the pseudo-Voit function of the film peaks (peaks **1–3**) and the substrate peak

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of the same composition, can differ significantly, since the films in the immediate vicinity of the interface with the substrate are always strained (in-plane tensile or compressive strain), which can, depending on the sign of the deformation, promote or counteract the formation of films enriched in REE, with an increased parameter in the plane of the interface. During the growth of $Tm_3Fe_5O_{12}$ on GGG a positive ϵ value, which means the in-plane tensile of the film, can lead to the growth of the film with an increased unit cell parameter, i.e. contribute to the formation of antisite defects $[RE_{Fe}]^{3+}$ and the formation of a layer of Tm-enriched garnet.

Based on the established sequence of growth of the interface layers, it is possible to assume a common cause and diffusion mechanism for their occurrence: the removal of gallium from the surface layer of the substrate causes a counter diffusion flow of iron ions into the substrate from the adjacent thin layer of stoichiometric Tm₃Fe₅O₁₂ garnet which leads to enrichment of the latter with a rare earth component. It should be noted that both interface layers have a garnet structure and therefore do not impair the epitaxial character and orientation of the deposited layer, considered at full thickness. Epitaxial nature of Tm₃Fe₅O₁₂ films on GGG(111) and YAG(111) substrates was confirmed by X-ray φ-scanning (Fig. 4a, b): since the position of the

reflections (800) of the film and substrate on the ϕ -scans coincided, it can be concluded that during $Tm_3Fe_5O_{12}/GGG(111)$ and $Tm_3Fe_5O_{12}/YAG(111)$ film growth, cube-on-cube heteroepitaxy occurs.

Garnet of the "correct stoichiometry" 3:5 on the diffraction pattern of the Tm₃Fe₅O₁₉/GGG sample corresponds to the peak number 1 (Fig. 3a, b). It corresponds to the reflection from the (444) plane, and its position corresponds to that expected based on the powder diffraction pattern (the position of the Tm₂Fe₅O₁₂(444) peak for the powder is indicated by a dashed line) taking into account the abovedescribed nature of epitaxial deformations: inplane tensile strain) and the resulting reduction in interplanar distances in the direction perpendicular to the substrate (out-of-plane). The fact that the deviation of the (444) reflex from the line for the powder is relatively small indicates a small amount of film deformation during heteroepitaxy on a substrate with a very different unit cell parameter. In accordance with theoretical concepts of heteroepitaxy [18], partial elimination of elastic stresses arising due to the parametric difference between the film and the substrate occurs in this case due to the emergence of misfit dislocations (MD) in the film. It is natural to assume that in this case the MD are formed at the boundary of the Tmexcess layer of Tm₃(Tm_xFe_{5-x})O₁₂ and stoichiometric layer Tm₃Fe₅O₁₂, which differ most significantly by the unit cell parameter.

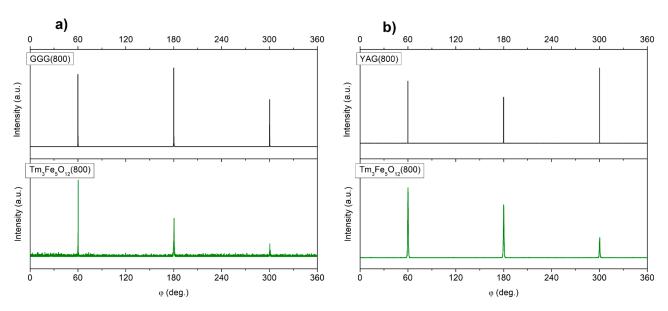


Fig. 4. X-ray φ -scans results for Tm₂Fe₅O₁₂ films on GGG(111) (a) and YAG(111) (b) substrates

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As the result of the deposition of Tm₃Fe₅O₁₂ films on a YAG(111) substrate, which has a lower unit cell parameter than the film and a larger parametric mismatch, was the formation of highly strained film, as indicated by the large deviation of the (444) reflex from its position for the powder sample (Fig. 3a). In this case, the stress is of compressive nature along the substrate, which, in accordance with the elastic nature of the deformation, led to stretching of the film along the normal to the substrate (as was indicated by the shift of the film peak towards smaller angles relative to the peak for the powder). Strong elastic strains in the film were manifested not only as a large shift of the (444) reflex, but also as its significant broadening: the width at half maximum (FWHM) was 0.31°. The magnitude of elastic strains s is of interest. The value of s was calculated based on the relative change of unit cell parameter of the film's e compared to a Tm₃Fe₅O₁₂ single crystal using the ratios:

$$\varepsilon = \frac{a_{\text{film}} - a_{\text{Tm}_5 \text{Fe}_5 \text{O}_{12}}}{a_{\text{Tm}_5 \text{Fe}_5 \text{O}_{12}}},$$
(6)

$$\sigma = \frac{E}{1 - \nu} \varepsilon, \tag{7}$$

where $a_{\rm film}=12.391$ Å (the unit cell parameter determined based on XRD data for the ${\rm Tm}_3{\rm Fe}_5{\rm O}_{12}$ film) and $a_{{\rm Tm}_3{\rm Fe}_5{\rm O}_{12}}=12.327$ Å (the unit cell parameter for the ${\rm Tm}_3{\rm Fe}_5{\rm O}_{12}$ single crystal), $E-{\rm Young's}$ modulus, ${\rm v-Poisson's}$ ratio. Since the elastic constants for ${\rm Tm}_3{\rm Fe}_5{\rm O}_{12}$ are not available in the literature, we used E=187 GPa and ${\rm v}=0.29$ for yttrium iron garnet ${\rm Y}_3{\rm Fe}_5{\rm O}_{12}$ for the calculation [19]. The compressive stresses determined in this way was 1.4 GPa. It is obvious that, despite its large value, the compressive stresses characterizes only the stress remaining after part of it has been eliminated by the appearance of MD at the film/substrate interface. The value of the theoretically possible elastic stress arising in the absence of the dislocation relaxation mechanism was estimated according to the formula:

$$\sigma = \frac{E}{1 - v} \cdot \frac{a_{\text{Tm}_3\text{Fe}_5\text{O}_{12}} - a_{\text{YAG}}}{a_{\text{YAG}}}$$
(8)

and composed ~7 GPA (a_{YAG} = 12.010 Å). Thus it becomes clear that the elastic stresses at the interface Tm₃Fe₅O₁₂ with YAG substrate by» 80%

are eliminated by mismatch dislocations.

In conclusion of the discussion of the X-ray diffraction results, it is necessary to state the degradation of $Gd_3Ga_5O_{12}$ -based substrates during gas-phase deposition, which led to the formation of additional phases in the composition of the $Tm_3Fe_5O_{12}$ film and to the impossibility of using GGG(111) substrates to obtain iron garnet films for applied research, including for studying their behavior in the terahertz region. In this regard, YAG substrates, which are quite stable under high-temperature conditions of film deposition using vacuum methods, despite the greater parametric mismatch, show a significant advantage.

The Raman scattering studies of synthesized ${\rm Tm_3Fe_5O_{12}}/{\rm YAG}(111)$ film and YAG substrates were conducted at wavelengths λ = 532 nm (Fig. 5a) and λ = 785 nm (Fig. 5b). The iron garnets belong to the space group Ia3d and according to [20] have the following set of vibrational modes:

$$\Gamma = 3A_{1g} + 6E_{g} + 9T_{2g} + 9T_{1u}.$$
 (1)

The YAG belongs to the same space group, and as studies of the YAG structure have shown [21], yttrium ions have 24 c-bonds with dodecahedral coordination, and aluminum ions have 16 a-bonds with octahedral coordination and 24 d-bonds with tetrahedral oxygen coordination. These 160 atoms (metals and oxygen) per unit cell create 18 phonon branches (17 optical and 1 acoustic T_{1u}). It should be noted that 18 Ramanactive modes A_{1g} , E_{g} and T_{2g} , and 9 IR-active oscillations T_{1u} are possible for the space group Ia3d [22].

Our Raman study showed that the best resolution of vibrational modes of the Tm₃Fe₅O₁₂ film against the background of a single-crystal YAG substrate was observed at $\lambda = 532$ nm (Fig. 5a): the spectrum allows modes at 117 (T_{2a}) , 182, 469, 650, 1114, 1284, 1817, 1910 and 1960 cm⁻¹. With less energy irradiation with a longer wavelength $\lambda = 785$ nm, weak differences were observed between the spectrum of the YAG substrate and the spectrum of the film (Fig. 5b): only 2 vibrational modes were allowed at 116 ($T_{2\sigma}$) and 251 cm 1 (E_{2g}). This is due to the fact that with a decrease in wavelength, starting from 800 nm, the absorption of rare earth iron garnets increases significantly. This, in turn, was associated with intense interband transitions and charge transfer

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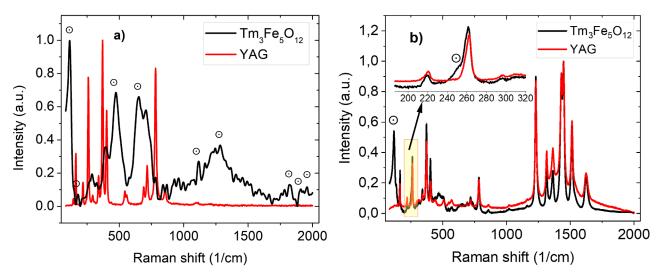


Fig. 5. Raman spectra of $Tm_3Fe_5O_{12}$ film on YAG(111) substrate at wavelengths of 532 nm (a) and 785 nm (b). \odot – vibrational modes related to the $Tm_3Fe_5O_{12}$ film

transitions from O^{2-} to Fe^{3+} or to the rare earth element ion RE^{3+} [23]. The obtained Raman shifts 116 and 251 cm⁻¹ were consistent with the data for $Tm_3Fe_5O_{12}$ films with the thickness of 20–300 nm on a substituted gallium-gadolinium garnet ($Gd_{2.6}Ca_{0.4}Ga_{4.1}Mg_{0.25}Zr_{0.65}O_{12}$) substrate with orientation (111), in the Raman spectra of which 2 modes were observed – 130 and 250 cm⁻¹ when shooting up to 300 cm⁻¹ was performed [24].

The study [22] demonstrated that Raman spectra have a relatively low signal-to-noise ratio at frequencies below a few terahertz (150 cm⁻¹). In this regard, as another method of vibrational spectroscopy, we performed THz-pulse spectroscopy, which allows restoring the dielectric characteristics of various materials in the frequency range from 0.1 to 10 Thz. In this study, the dielectric characteristics of GGG and YAG substrates (with thicknesses of 513 µm and 502 µm, respectively) were investigated based on measurements of THz radiation transmission (Fig. 6). These measurements also indicate that the YAG(111) substrate is more promising than the GGG(111) substrate, since its characteristic spectrum does not show significant absorption in a wider frequency range from 0.25 to 3.0 THz, while GGG substrates have an absorption line at a frequency of 2.5 THz, which reduces the spectral range for film analysis to 0.2–2.2 THz. Thus, despite the stressed nature of the Tm_zFe_zO₁₀ films, obtained on a YAG(111) substrate, such films appear more promising for further terahertz

research. A resonant absorption mode was detected for the YAG(111) substrate at a frequency of 3.25 THz (108 cm⁻¹). The observed resonant absorption mode can be explained by a two-phonon difference process, as a result of which an acoustic phonon is transferred to the optical branch [22].

Due to the low film thickness compared to the wavelength of THz radiation, we simultaneously measure THz fields reflected and transmitted through the ${\rm Tm_3Fe_5O_{12}}$ film on the YAG substrate. This approach improves the accuracy of the reconstruction of THz characteristics of thin films [25, 26]. The transmittance/reflectance spectra of the YAG(111) substrate and ${\rm Tm_3Fe_5O_{12}}$ film on

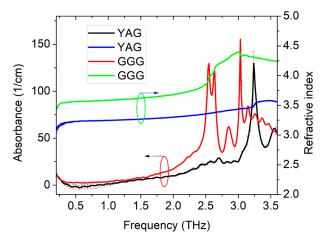


Fig. 6. Absorption coefficient and refractive index spectra of GGG(111) and YAG(111) substrates in the frequency range from 0.2 to 3.6 THz

this substrate are shown in Fig. 7a. As a result of the reconstruction of the THz characteristics [26], the spectra of the absorption coefficient and refractive index of the thulium iron garnet film on the YAG substrate were obtained (see Fig. 7b). The absorption coefficient tends to increase slowly in the high frequency region. The refractive index of the film has a large dispersion. At a fixed frequency of 2 THz, the absorption coefficient of the film was 13 cm⁻¹, and the refractive index was 2.8, which is typical for materials with low conductivity.

4. Conclusions

 $Thin\,Tm_{_{3}}\!Fe_{_{5}}O_{_{12}}iron\,garnet\,films\,were\,obtained$ on single-crystal GGG(111) and YAG(111) substrates using the MOCVD method. It was found that due to the volatility of gallium(I) oxide under deposition conditions (910 °C and $P_{O2} = 1$ mbar) the introduction of iron oxide into the surface layer of the substrate and enrichment of the adjacent layer of the film with thulium oxide occurs, which leads to the formation of non-stoichiometric garnet Tm₃(Tm₃Fe₅₋₃)O₁₂ with antisite defects [Tm_{Fe}³⁺]. Thus, the epitaxial film of stoichiometric garnet Tm₃Fe₅O₁₂ grows on a GGG substrate with two interface layers. On the YAG substrate, the thulium iron garnet film is deposited without chemical interaction at the interface, but experiences strong inplane compressive stress from the substrate. Films on a YAG substrate, transparent up to a frequency of 3.5 THz, have an advantage over films on GGG from the point of view of potential application in THz-range technology. As a result of the reconstruction of the THz characteristics, the frequency dependences of the absorption coefficient and the refractive index of the Tm₃Fe₅O₁₂ film on a YAG substrate were obtained.

Author contributions

Markelova M. N. – scientific supervision, research concept, processing of X-ray diffraction results and text writing. Hafizov A. A., Shi X. – conducting experiments on gas-phase deposition of thin films, processing the results of X-ray diffraction. Graboy I. E. – development of methodology, text writing. Shanin M. S., Konnikova M. R. – conducting experiments on Raman spectroscopy and terahertz spectroscopy. Shkurinov A. P. – scientific supervision, research concept, methodology development. Kaul A. R. – scientific supervision, research concept, methodology development, text editing.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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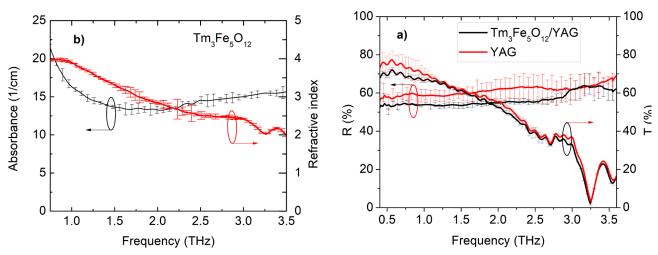


Fig. 7. Transmission, reflection (a), absorption and refractive index spectra of $Tm_3Fe_5O_{12}$ film (b) on YAG substrate in the THz frequency range. The standard deviation is calculated for three independent averages of 1000 accumulations each

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Topography and microrelief electroless Ni-P coatings at different loading densities

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Abstract

The aim of this study was to investigate the growth processes of Ni-P coatings at different loading densities of the electroless nickel plating bath. The Ni-P coatings can be used to improve the thermal and corrosion resistance of optical fiber used to manufacture sensors of various physical quantities.

When depositing coatings on optical fiber, the loading density is an important parameter. The study investigated the influence of loading density on the topography, microrelief, roughness, and growth mechanism of Ni-P coatings using non-contact high-resolution optical profilometry. An increase in loading density from 0.5 to 3.0 dm²/l did not lead to a significant change in the roughness parameters of the coatings. During the growth of coatings, "spheroids" strongly elongated in the plane of the substrate were formed on the surface. An analysis of growth processes was carried out within the framework of the layered growth mechanism. The growth rates of spheroids in normal and lateral directions were assessed. An increase in the loading density led to a decrease in the growth of spheroids in the normal direction, while the growth rate in the lateral direction was maximal at a loading density of 2 dm²/l. Statistical analysis of the sizes of the "spheroids" showed that their distribution deviated from the normal law, which may be due to the fact that not only individual spheroids, but also their aggregates were considered in calculations. Another reason for the deviation may be a decrease in the proportion of the active surface on which the coating is deposited as the loading density increases.

Since increasing the loading density from 0.5 to 3.0 dm 2 /l did not lead to a significant change in the roughness parameters even with coating thickness of ~ 8 µm or higher; deposition of coatings with the thickness up to 3.5 µm thick on optical fibers can be carried out at the specified loading densities with acceptable surface roughness.

Keywords: Optical fiber, Ni-P coatings, Electroless deposition, Non-contact profilometry, Topography, Microrelief, Roughness, Growth mechanism, "Spheroids", Statistical analysis

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1. Introduction

Chemically electroless Ni-P coatings have a variety of applications in mechanical engineering, power engineering, and electronics [1] due to the ability to deposit hard, corrosion-resistant, heat-resistant coatings on products of complex configuration and also on dielectrics [2–6].

Currently, optical fiber (OF) is used not only for transmitting information, but also for the production of sensors of various physical quantities (temperature, stress, pressure, etc.) in the oil, gas, and nuclear energy industries [7–10]. Operating conditions of sensors may include temperatures up to 600 °C and aggressive environments. Usually, the OF is produced with a polymer protective and strengthening coating, which protects it from external mechanical influences and moisture [11–12]. Such optical fiber can operate at temperatures up to 200 °C [13]. In order to increase the operating temperature, metallized OF are used, in which a protective and strengthening coating of copper or aluminum is applied directly in the drawing column from molten metal. The operating temperature of such fibers in air is 350–400 °C. However, the copper coating does not sufficiently protect the optical fiber at high temperatures [14]. Nickel-coated optical fiber has higher corrosion resistance [15].

A layer of Ni-P coating can be applied over the metallized copper layer to increase corrosion resistance and heat resistance [16–17]. Such coatings were used in the manufacture of pressure sensors [18]. The coating can be two-layer, a thin layer of Ni-P coating and a thicker layer of electrodeposited nickel [19].

The roughness of the coatings [19–21], determined by the deposition conditions and the composition of the solution, is of great importance. In addition, the lower the roughness of the coatings, the less probable the formation of cavities due to stresses arising from temperature differences [20].

Thus, the application of nickel coatings to metallized optical fiber can be carried out in two ways: electrochemical or electroless nickel plating. In the first case, continuous pulling of the fiber through the nickel plating bath will be required; in the second case, due to the peculiarities of the process, the coating can be applied to a spool of optical fiber. The influence

of a number of parameters on the microrelief and growth processes of Ni-P coatings was previously studied [22–25]. In electroless deposition, one of the significant factors is the loading density (the ratio of the area of the surface to be coated to the volume of the solution). In this study, the influence of loading density on the roughness and microrelief of Ni-P coatings was investigated using polished steel samples as the model objects.

2. Experimental

The coatings were deposited on polished samples of 20X13 steel and optical fiber with a quartz core diameter of 125 μ m and a copper layer thickness of 18 μ m. A copper layer was deposited on the optical fiber from the melt directly in the fiber drawing column.

Mechanical polishing of steel was carried out using a STRUERS LaboPol-2 polishing machine. The coatings copy the microrelief of the substrate, therefore, to eliminate the influence of the microrelief of the substrate on the microrelief of the deposited coatings, careful polishing of the substrate was performed.

Initially, polishing was carried out on Sic-Paper grinding wheels of class 220÷1000, which corresponds to a grain size of 50÷10 μ m. Next, DiaPro diamond suspensions with particle sizes from 9 to 1 μ m (Struers) were used. For the deposition of coatings, only steel samples with R_a (roughness) not exceeding 4 nm were selected.

Coatings were deposited on 20X13 steel samples for the investigation of the topography and microstructure of Ni-P alloys. The size of the samples was determined by the loading density, which varied from 0.5 to 3.0 dm²/l.

Before deposition, the samples were degreased with alcohol, washed with distilled water and pickled in HCl (1:1) for 30–60 s. Deposition was carried out at 358 K under thermostatted conditions from a solution of the following composition, (M): $\mathrm{NiCl_2} - 0.12$; $\mathrm{NaH_2PO_2} - 0.114$; $\mathrm{CH_3COONa} - 0.12$. The temperature was maintained with an accuracy of $\pm 1^\circ$, the pH of the solution was 4.5. The reagents for the preparation of solutions were of chemically pure and analytical reagent grade ($\mathrm{NaH_2PO_2}$) and were recrystallized twice. The required pH value was set by adding HCl (extra pure grade).

The thickness of Ni-P coatings on optical fiber was 3.5–4.0 µm. The plasticity of the optical fiber decreased with a higher coating thickness.

The coating thickness was determined gravimetrically, taking the coating density for Ni-P deposits to be 8.0 g/cm³ [1].

The growth of the coatings and their phosphorus content were studied using a Hitachi S3400N scanning electron microscope and New View-7300 non-contact profilometer (Zygo). The following parameters were analyzed: PV – the maximum difference in height between the highest and lowest points of the profile surface; $R_{\rm A}$ - roughness, rms – standard deviation from the central line, R_z – the average absolute value of the five highest peaks and the five deepest depressions. The phosphorus content in the coatings was determined using energy dispersive analysis.

In addition, the average radii (\overline{r}), heights of "spheroids" (h) and the average ratio of these

were determined directly from the

obtained microprofiles. The analysis of the obtained results was carried out using the MS Excel package. The confidence probability was adopted equal 0.95 for calculations of confidence intervals.

3. Results and discussion

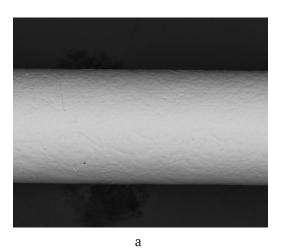
Electroless Ni-P coatings on the surface of the optical fiber must be uniform (Fig. 1a), with a smoothed microstructure and low roughness, which in turn is determined by the deposition conditions, the composition of the electroless nickel plating solution and the loading density. The microrelief of the coatings is determined by the presence of "spheroids" on the surface (Fig. 1b). These are formations elongated in the plane of the substrate with a fairly high ratio of the radius to the height of the "spheroid" [22–23].

The usually used loading density for deposition of coatings does not exceed 2.0 dm²/l [1]. The study considers the influence of this technological parameter on the growth processes of Ni-P coatings.

The deposition rate had a steady tendency to decrease with an increased loading density (Table 1), which was described earlier [1].

The deposition rate decreased almost twofold with an increase in loading density from 0.5 to 3.0 dm²/l (Table 1). Starting from a loading density of 1.0 dm²/l the decrease in deposition rate with increasing loading density occurred almost linearly. An increase in the loading density caused a rapid decrease in the pH of the electroless nickel plating solution (Table 1), a decrease in the concentration of nickel ions and the reducing agent, thus the deposition almost stopped after some time. This was evidenced by a comparison of the dependences of the coating thickness obtained with deposition times of 40 and 60 min (Fig. 2). At a loading density of 3.0 dm²/l the thickness of the coatings were almost identical.

The substrate had a significant influence on the topography of the growing coatings, since



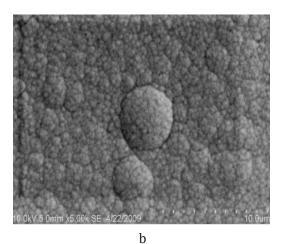


Fig. 1. (a) Micrograph (250x magnification) of Ni-P coating on the surface of copper protective layer on the optical fiber, and (b) "spheroidal" structure of coating (5000× magnification)

Table 1. Coating deposition rate, phosphorus content and change in pH value of electroless nickel plating solution (Δ pH) after deposition at different loading densities

Loading, dm ² /l	Deposition rate, µm/h	Phosphorus content, wt. %	∆рН
0.5	13.6±1.2	8.1±0.4	0.24±0.06
1.0	13.1±0.5	8.7±0.2	0.39±0.04
1.5	11.9±0.6	9.4±0.2	0.48±0.05
2.0	9.9±0.5	9.8±0.3	0.72±0.07
3.0	7.5±0.5	10.7±0.4	0.85±0.04

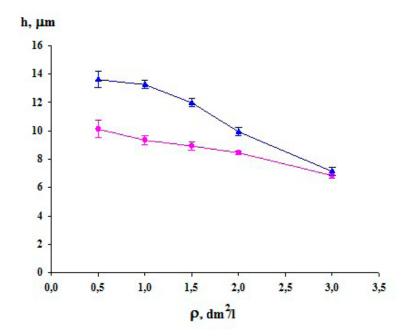


Fig. 2. Dependence of the coating thickness on the loading. Deposition time, minutes: 40 (●), 60 (▲)

Ni-P coatings accurately copied all the details of the substrate surface relief. The decrease in the deposition time will result in a reduction in the coating thickness, which may cause the substrate microrelief to affect the coating's surface topography. Taking this into account, the deposition duration in further studies was 60 min. Deposition at minimum loading density (0.5 dm²/l) led to the formation of a typical surface relief of Ni-P coatings (Fig. 3). The surface of the coating contained numerous "spheroids", most of which overlapped into agglomerates of various shapes. At the same time, the surface relief was quite smooth, since the height of most "spheroids" did not exceed 30-40 nm.

Increase in the loading density did not qualitatively change the surface topography of the coatings (Fig. 4). The number of "spheroids" on the surface of the coating increased, they

become smaller and formed agglomerates of various shapes. For this reason, the detection of clearly distinguishable spheroids was a difficult task.

Individual pores on the surface were detected on the surface, probably formed as a result of the adsorption of hydrogen bubbles, which prevented the deposition of the coating in this area.

The formation of grooves with a linear shape on the surface can probably be associated with the paths of hydrogen bubble removal from the coating surface. Another reason for their occurrence may be insufficient polishing of the steel substrate.

Increase in loading density up to 2 dm²/l led to an increase in the radii of the "spheroids", which can be concluded from a comparison of the micrographs (Figs. 4 and 5). At the same time, the height of the "spheroids" became

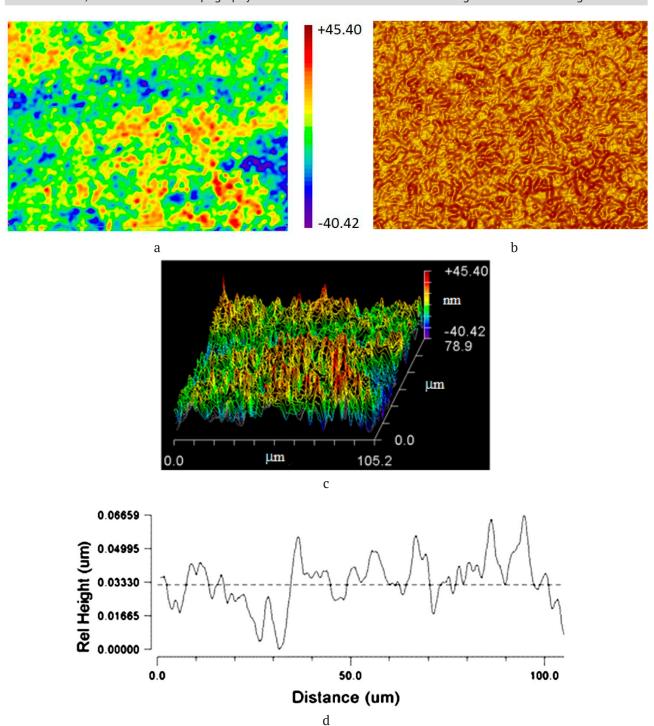


Fig. 3. Topography and microrelief of the coating formed at loading of $0.5 \, \text{dm}^2/\text{l}$: a - topographical height map; b - map of gradient; c - 3D image of the surface; d - microprofile along the selected direction

smaller. The topography and microrelief of the growing coating did not change fundamentally at the maximum loading density in our series of experiments of 3.0 dm²/l; in addition to the "spheroids", the same characteristic structural elements – pores and linear grooves.

For a more detailed comparison of roughness, the roughness parameters of the coatings obtained at the studied loading densities were analyzed (Table 2).

The comparison of the microroughness parameters of the coatings indicates that with an

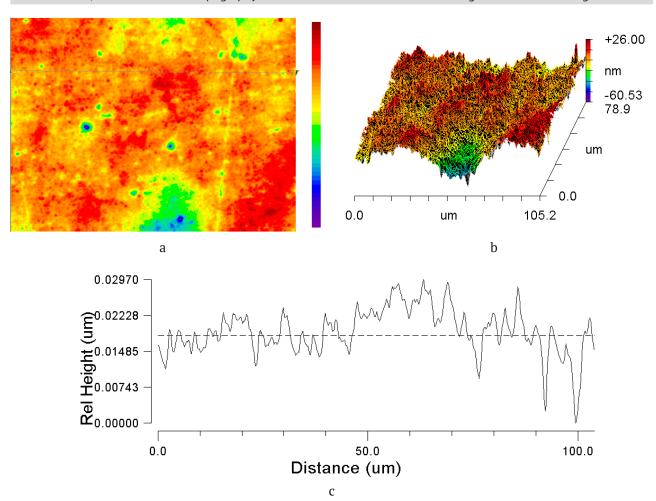


Fig. 4. Topography and microrelief of the coating formed at loading of 1 dm²/l: a – topographical height map; b – 3D image of the surface; c – microprofile along the selected direction

 Table 2. Parameters of microroughness of Ni-P coatings, formed at different loadings

Loading, dm ² /l	R _a , nm	rms, nm	$R_{\rm z}$, nm	PV, nm
0.5	8.5±2.1	10.8±2.6	54.7±23.9	81.1±17.4
1.0	7.1±1.4	9.3±1.5	62.5±4.9	95.6±12.6
2.0	4.7±1.2	6.2±1.6	26.8±1.3	50.3±7.4
3.0	5.2±1.3	7.4±0.3	51.3±16.6	72.5±25.0

increase in the loading density, the roughness of the coatings did not tend to increase. Moreover, the parameters of microroughness of the coatings obtained at a loading density of 2.0 dm²/l were minimal in the studied series of samples.

At a loading density of 3.0 dm²/l, the microroughness parameters were also comparatively low. Some increase of parameters was possibly due to the fact that at a given loading density, localization of coating growth occurred at the end of deposition. This caused the appearance of irregularly shaped formations on the coating surface with a height of ~100 nm, protruding

above the rest of the smoothed surface (Fig. 6). The approximate diameter of such formations was $100-250\,\mu m$.

The reason for the occurrence of such formations may be due to the low rate of the deposition process, the surface of the coating does not have time for renewal, and is poisoned by impurity compounds adsorbed from the solution. Only separate areas of the growing surface remain active, which causes localization of the growth of the coating.

Since the process of coating growth occurred through the formation of "spheroids", further treatment of the results consisted of finding the

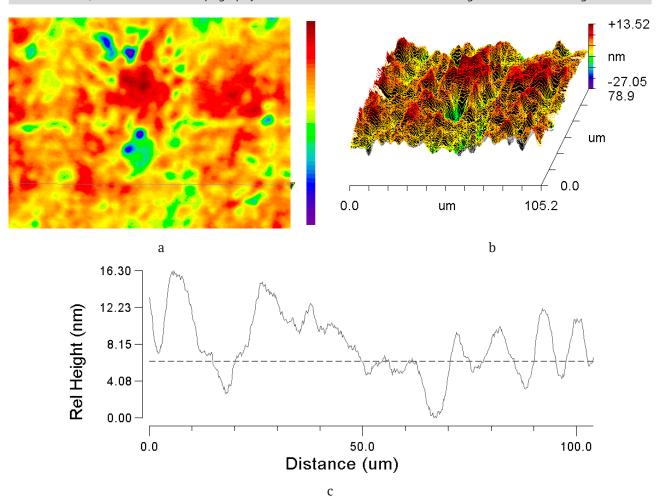


Fig. 5. Topography and microrelief of the coating formed at loading of 2 dm²/l: a – topographical height map; b – 3D image of the surface; c – microprofile along the selected direction

average values of the height (h), the radius of the "spheroids" (\overline{r}) and the relationship of these parameters $\overline{\left(\frac{r}{h}\right)}$ (Fig. 7) based on profilograms.

For processing fragments of profilograms where the "spheroids" clearly did not overlap were used.

The average values for each of the loading densities were obtained by averaging several samples, each of which yielded from 70 to 200 experimental values. The results of the treatment are presented in Table 3.

As can be seen from Table 3, changes of the parameters \overline{h} , \overline{r} , and the $\overline{\left(\frac{r}{h}\right)}$ ratio were non-monotonic. With increasing loading density up to 1 dm²/l the radii and heights of the "spheroids" decreased, while the $\overline{\left(\frac{r}{h}\right)}$ ratio practically did

not change. The latter circumstance indicates a similar nature of growth of the coatings. At the same time, the decrease in the average radii and heights of the "spheroids" may be associated with an increase in passivation phenomena during the growth of the coating. A higher loading density caused more intensive acidification of the near-electrode layer of the solution, a decrease in the deposition rate, and increased inhibition of the deposition process by products.

Further increase in loading density up to 2.0 dm²/l led to a noticeable increase in the radii of the "spheroids" with a slight decrease in their heights. These opposite trends in the change of the parameters of the "spheroids" significantly

increased the
$$\left(\frac{r}{h}\right)$$
 ratio.

One of the probable reasons for the increase in the radii of the "spheroids" may be the insufficient I. V. Petukhov, N. A. Medvedeva Topography and microrelief electroless Ni-P coatings at different loading densities

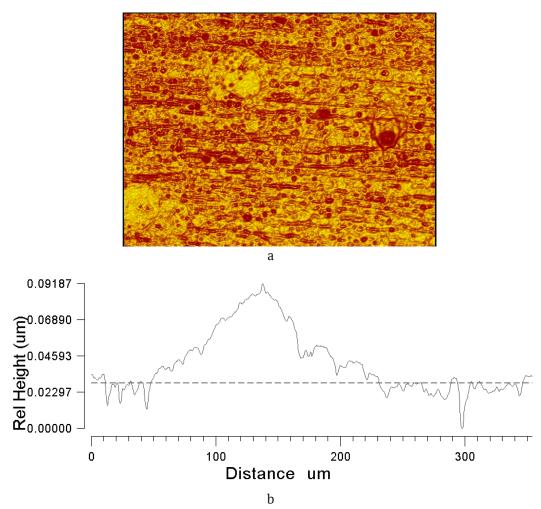


Fig. 6. Topography and microrelief of Ni-P coatings obtained at the loading of $3.0 \, \text{dm}^2/\text{l}$: a – map of gradient; b – microprofile along the selected direction

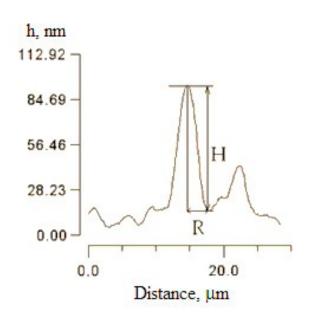


Fig. 7. The scheme of profilogram treatment

resolution of interference microscopy and 3D profilography methods for the study of the fine structure of the resulting "spheroids". It is possible that the "spheroids" observed under these conditions were aggregates of smaller "spheroids". The complex structure of "spheroids" was indicated by the results obtained by the AFM method [25–27]. At the same time, it cannot be ruled out that under conditions of sufficiently low catalytic activity of coating and a low deposition rate, the only possibility for growth was growth in the horizontal direction, which allows the generation of a juvenile coating surface. Similar phenomena were observed during the growth of coatings in the presence of stabilizing additives [23].

A decrease in the radii and the $\left(\frac{r}{h}\right)$ ratio with a slight change in the height of the "spheroids"

Table 3. Parameters determined the size of "spheroids" and their growth rates in vertical plane (V_v) and substrate plane (V_s)

Loading, dm ² /l	\overline{r} , μ m	\bar{h} , nm	$\overline{\left(\frac{r}{h}\right)}$	$V_{_{\Gamma}}$, nm/s	V _B , nm/s
0.5	2.77±0.14	21.22±1.46	143±9	880±46	6.74±0.44
1	1.53±0.08	11.1±0.7	155±9	994±59	6.40±0.01
2	3.63±0.26	8.1±0.7	478±29	2319±139	4.85±0.01
3	2.35±0.18	10.7±1.4	264±26	1073±107	4.05±0.01

was observed at a loading density of 3.0 dm²/l (Table 3).

Despite the complex nature of the change in the parameters of the "spheroids", we attempted to use the layered growth mechanism to describe their growth processes [22-23, 25]. Within the framework of this mechanism, the average growth rates of "spheroids" in the lateral (V_h) and normal directions (V_h) were determined (Table 3).

Analysis of the results shown in Table 3 indicates that increasing the loading density decreased the rate of normal growth. It should be noted that its increase from 0.5 to 1.0 dm²/l had a relatively weak effect on growth rates in both normal and lateral directions.

The main reason for the decrease in normal growth rate was the acidification of the electroless nickel plating solution during the deposition process (Table 1). The acidification of the surface layer of the solution was even more expressed [28]. These changes in the composition of the solution blocked vertical growth.

The decrease in normal growth rate with a decreasing pH of the solution has been described previously [23]. Acidification of the solution with increasing loading density should also lead to a decrease in lateral growth, but the results indicated an increase in the growth rate of the "spheroids" in this direction. This was probably due to the fact that in this case, using profilometric measurements, we recorded the dimensions of not a single "spheroid", but a formation composed by several "spheroids".

Further processing of the obtained results consisted of statistical analysis of samples of radii and heights of "spheroids", *r/h* ratio for each of the studied loading densities.

For each loading density, histograms of the spheroid size distribution were obtained (separately, histograms of the distribution by r and h), and the distribution of the ratio of these values (r/h). Histograms of the distribution of the values of r, h and the r/h ratio for coatings obtained at different loading densities are shown in Figs. 8–9.

In order to assess the compliance of the obtained distributions with the normal law, the values of the coefficients of asymmetry (γ_1^*), excess (γ_2^*) and the values of these coefficients acceptable for the normal distribution were calculated [28]:

$$\left|\gamma_{1}^{*}\right| \leq \gamma_{1,\text{norm}}^{*},\tag{1}$$

$$\left|\gamma_{2}^{*}\right| \leq \gamma_{2,\text{norm}}^{*},\tag{2}$$

Maximum permissible values of asymmetry coefficients $\left(\gamma_{1,\text{norm}}^*\right)$ and excess $\left(\gamma_{2,\text{norm}}^*\right)$ were calculated using the following formulas:

$$\gamma_{1,\text{norm}}^* = 3\sqrt{D(\gamma_1^*)}, \qquad (3)$$

$$\gamma_{2,\text{norm}}^* = 5\sqrt{D(\gamma_2^*)},\tag{4}$$

where $D(\gamma_1^*)$ and $D(\gamma_2^*)$ – the dispersions of these values determined using the following equations:

$$D\left(\gamma_{1}^{*}\right) = \frac{6\left(n-1\right)}{\left(n+1\right)\left(n+3\right)},\tag{5}$$

Table 4. Statistical parameters of spheroids' radius distribution

Loading, dm ² /l	r, μm	γ_1^*	γ_2^*	n	$\gamma_{1 \text{ normal}}^*$	γ _{2 normal}
0.5	2.8±0.1	1.7 6	0.81	142	0.60	1.94
1	1.5±0.1	0.78	1.02	195	0.52	1.68
2	3.6±0.3	-0.11	0.46	82	0.78	2.46
3	2.4±0.2	0.28	0.59	71	0.84	2.62

Table 5. Statistical parameters of spheroids' height distribution

Loading, dm ² /l	h, nm	γ_1^*	γ_2^*	n	γ* 1 normal	γ _{2 normal}
0.5	21.2±1.5	1.10	0.90	142	0.60	1.94
1	11.1±0.7	1.52	1.15	195	0.52	1.68
2	8.1±0.7	1.64	0.95	82	0.78	2.46
3	10.7±1.4	1.25	1.18	71	0.84	2.62

Table 6. Statistical parameters of r/h ratio distribution

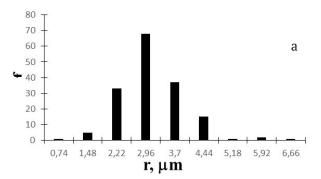
Loading, dm ² /l	r/h	γ_1^*	γ_2^*	n	$\gamma_{1 \text{ normal}}^*$	γ _{2 normal}
0.5	143±9	1.66	0.81	142	0.60	1.94
1	155±9	22.2	3.69	195	0.52	1.68
2	478±29	8.88	2.57	82	0.78	2.46
3	264±26	0.11	0.51	71	0.84	2.62

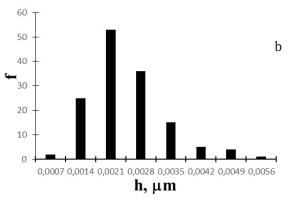
$$D(\gamma_2^*) = \frac{24n(n-2)(n-3)}{(n+1)^2(n+3)(n+5)}.$$
 (6)

The results of the calculations performed for samples obtained at different loading densities are shown in Tables 4–6.

A comparison of the asymmetry and excess coefficients calculated for a sample of experimental values of the "spheroid" parameters with similar coefficients for a normal distribution allows to conclude that for most loading densities, the distribution of "spheroids" by size does not obey the normal law. Deviations from the normal law are caused by an increase in asymmetry coefficient values. The most heterogeneous is the distribution of the parameter *r/h*. This is obvious even from the appearance of the histograms (Fig. 8c, 9c) without additional statistical processing. The excess coefficient for most cases corresponded to a normal distribution.

The fact that the distribution of "spheroids" by size corresponded to the normal law may indicate that all "spheroids" grow by the same mechanism. The possible simultaneous implementation of normal and layered growth mechanisms was mentioned in [24]. Under the deposition conditions used in the study, no "spheroids" with a low r/h ratio were observed, which is typical for the mechanism of normal growth. Therefore, deviations from the normal distribution may be associated with the fact that in a number of cases, instead of individual spheroids, we registered aggregates of these "spheroids". The used method did not provide the required horizontal resolution. In order to obtain the required resolution, atomic





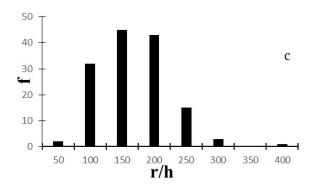
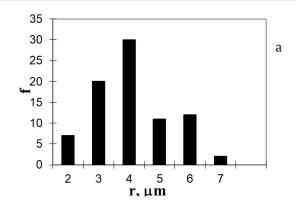
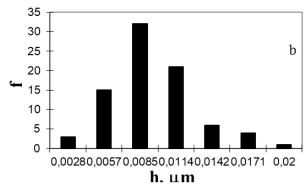


Fig. 8. r (a), h (b) and r/h ratio (c) distributions for spheroids obtained at loading of 0.5 dm²/l





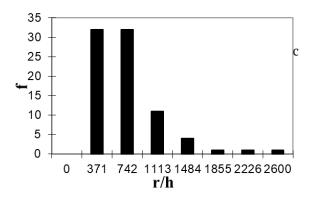


Fig. 9. r (a), h (b) and r/h ratio (c) distributions for "spheroids" obtained at loading of 2.0 dm²/l

force microscopy should be used, but due to the labor intensity of the method, it will be difficult to obtain a similar sample size. The presence of individual "spheroids" and aggregates of them in one statistical sample may lead to deviations from the normal law.

At the same time, the obtained results do not contradict the layered growth mechanism, since during the coating deposition process, the pH of the electrolyte changes, the deposition rate decreases, which can enhance passivation phenomena on the growing surface. Due to this, the movement of individual atomic layers spreading over the surface can be blocked. For

this reason, multiple growth centers (several growth layer packages) arise. These newly formed "spheroids" grow until they overlap with each other (Fig. 1b). In this way the structuring of the growing "spheroid" occurs. The rounded shape of these formations is determined by the fact that at this shape the surface energy is minimal at the spheroid-solution interface, since the amorphous structure of the coatings does not require crystalline faceting. If there is no change in the growth mechanism, it can be assumed that this will not have a significant impact on the calculation results presented in Table 3.

4. Conclusions

Thus, the investigation of growth processes at loading densities from 0.5 to 3.0 dm²/l allows us to conclude that an increase in the loading density in the studied range of values does not lead to an increase in the roughness of the coatings. Probably, the condition for the transition to the normal growth mechanism and the corresponding increase in the roughness of the coatings is a sufficiently high catalytic activity of the surface accompanied by the discharge of nickel ions in the diffusion or diffusion-kinetic mode, which can occur with an increase in the temperature of the electroless nickel plating solution or with a decrease in the concentration of nickel ions [24].

An increase in the loading density increases the acidification of the electroless nickel plating solution during the deposition process, which contributes to an increase in the phosphorus content in the coatings and, as a consequence, a decrease in their catalytic activity in the oxidation reaction of the reducing agent (sodium hypophosphite), due to which the deposition rate significantly decreases.

Under condition of low deposition rates (low catalytic activity of coatings), conditions for the discharge of nickel ions in diffusion mode and the transition to the normal growth mechanism with the formation of dendritic deposits probably are not created. Therefore, with an increase in loading density, coatings with a smooth surface can be formed.

Some additional increase in roughness occurs only when the deposition rate is reduced to almost zero and the growth process occurs only on individual active areas of the coating surface. The obtained results should be taken into account when depositing Ni-P coatings on the surface of optical fiber.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Coulometric analysis method for determining the concentration and degree of oxidation of vanadium in the electrolyte of a vanadium flow battery using a hydrogen vanadium cell

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Abstract

Determining the vanadium content and the average degree of oxidation of vanadium ions in an electrolyte is a highly important task, both in the production and operation of vanadium flow batteries and in scientific research aimed at improving the performance characteristics of electrolytes throughout their entire life cycle. This article proposes a solution to this issue using the coulometric analysis of electrolyte samples circulating through a cell with a membrane-electrode unit consisting of a gas diffusion hydrogen electrode, a proton exchange membrane, and a liquid flow electrode. The coulometric analysis involves the oxidation of the sample to the highest degree of vanadium oxidation with further reduction to an oxidation state of +4. The parameters of the procedure (polarization modes and completion conditions) were chosen in order to minimize the relative error in determining the concentration of vanadium up to 5% and the average degree of oxidation up to 2% based on model composition electrolytes with different concentrations and degrees of vanadium oxidation, including sulfuric acid, as well as mixed acid ($H_2SO_4 + HCl$) compositions.

Keywords: Vanadium electrolyte, Vanadium flow batteries, Coulometry, Hydrogen electrode, Concentration, Degree of oxidation of vanadium

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1. Introduction

Redox flow batteries are a promising electrochemical technology for integrating with renewable energy sources due to their flexibility and scalability [1-3]. Vanadium flow batteries (VFB) have become obvious competitors in the future energy industry as dozens of projects have been implemented around the world, and their integration into energy networks continues [4, 5]. Vanadium flow batteries (VFB) consist of electrolyte tanks, graphite electrodes, ion exchange membranes, and circulation systems to ensure efficient energy storage. A significant part (from one third to one half) of the cost per kilowatt-hour (kWh) of energy accumulated using this technology is accounted for by electrolytes based on vanadium compounds [6]. Numerous studies have been aimed at optimizing the composition of vanadium electrolytes and their preparation methods, as well as their regular monitoring and performance improvement. The methods of electrolyte production and the study of reactions occurring in it are being constantly developed and improved due to continuous research [7–11].

The crossover of vanadium compounds [12, 13] and the occurrence of side electrode reactions during the operation of the VFB [14-16] result in decreased capacity due to a violation of the charge balance between positive and negative electrolytes (with the same volume of positive and negative electrolytes, the same concentration of vanadium compounds in them, and the total average degree of oxidation of the two electrolytes around +3.5). To solve this problem, various approaches are being developed, including passive and active rebalancing of the positive and negative capacity of the electrolyte: hydraulic shunting [17], pressure gradient [18], introduction of reducing agents [19], and auxiliary electrochemical devices that restore a vanadium electrolyte using water [20, 21] or hydrogen oxidation [22].

It should be noted that the production of an electrolyte and its operational maintenance require regular monitoring of the key characteristics of the electrolyte: the concentration and average degree of oxidation of vanadium ions. Several methods were proposed to solve this problem. For example, in [23–

25] it was proposed to monitor the average oxidation state of a vanadium electrolyte using potentiometric measurements. To determine the total vanadium content in an electrolyte sample, another analytical method should also be used. In [26] it was proposed to add a reducing or oxidizing agent to transfer vanadium to a required valence state, while the concentration of vanadium in each valence state was calculated based on the volume of the initial sample and the amount of oxidizing/reducing agent added. The method can be supplemented by measurements of the absorption spectrum of the obtained solutions. The spectrophotometric method described in [27, 28] also requires preliminary calibration using references. The concentrations of vanadium compounds in the oxidation states of +3 and +4 in the test sample were determined based on these calibration dependencies. For electrolytes with a concentration of 1-2 M, the optical absorption ceased to depend linearly on the concentration, which complicated the calibration process. In addition, changes in the electrolyte composition regarding concentration and acid composition could cause shifts in the characteristic absorption peaks of vanadium compounds in the oxidation states of +3, +4, and +5 due to changes in the acidity of the solution and the appearance of new absorption bands caused by concentrationdependent complex vanadium compounds [29, 30]. The use of microelectrodes for amperometric studies seems promising, but its applicability was demonstrated only for reversible redox pairs of low-concentration vanadium electrolytes [31, 32].

The purpose of this study was to evaluate the applicability of the coulometric method for determining the characteristics of an electrolyte using a flow cell with a membraneelectrode assembly (MEA) of the composition H_2 (PtC)//proton exchange membrane// $V^{+(n-1)}/V^{+n}$ composition, where *n* ranged from 3 to 5. Previously, MEA of similar composition (combining gas diffusion hydrogen and liquid flow vanadyl/ vanadate half-cells) were proposed for autonomous energy storage systems [33, 34] and as auxiliary devices for electrochemical rebalancing [35]. As far as we know, such MEA have not been used for analytical purposes in relation to the composition of the VFB electrolyte [36-38].

2. Experimental

Three samples were tested to determine the degree of oxidation and concentration of vanadium ions: 1- sulfuric acid electrolyte with a composition of $1\,\mathrm{M\,VOSO_4}$ in $4\,\mathrm{M\,H_2SO_4}$ (vanadium oxidation state +4), 2- sulfuric acid electrolyte with a concentration of 1.6 M vanadium and an average degree of vanadium oxidation of +3.5 in $2.4\,\mathrm{M\,H_2SO_4}$ and 3- mixed acid electrolyte from a commercial vanadium flow battery.

For sample 1, we used a sulfuric acid electrolyte prepared from vanadyl sulfate VOSO₄·3H₂O trihydrate of the reagent (Reakhim, Russia) and sulfuric acid (95%, Sigmatek, Russia) of the reagent grade. The vanadyl sulfate portion was dissolved in a pre-prepared 4 M sulfuric acid solution until a predetermined solution volume was reached.

For sample 2, we prepared a sulfuric acid electrolyte with a known vanadium concentration (1.6 M) and an average degree of vanadium oxidation of +3.5 in 2.4 M sulfuric acid from vanadium pentoxide V2O5 99.6% (Sigma-Aldrich, USA). The chemical reduction of vanadium pentoxide V_2O_5 to an oxidation state of +4 was conducted using oxalic acid (dihydrate, RG, VWR Chemicals, USA) in a 2.4 M solution of sulfuric acid with further electrolysis of a sulfuric acid solution of vanadium with an oxidation state of +4 divided into two portions to oxidation states of +3 and +5, respectively [39].

Sample 3 was a commercial mixed acid electrolyte synthesized according to the instruction received from LIAONING GREPALOFU NEW ENERGY CO.LTD within the ranges of precursor values provided by the company: vanadium (III) sulfate content -10-15 wt. %, vanadium (IV) sulfate -8-12 wt. %, sulfuric acid -10-12 wt. %, water -50-60 wt. %, hydrochloric acid -10-12 wt. %. According to this instruction, the average concentrations in sample 3 were as follows: $C_v = 1.65$ M, $C(SO_4) = 2.7$ M, C(CI) = 3.014 M. In this case, the concentration of vanadium in a commercial sample can be in the range from 1.32 to 1.98 M.

A background electrolyte solution was also prepared – a solution of sulfuric acid with a concentration of 4 M.

Electrochemical measurements were performed on the installation, the scheme of

which is presented in Fig. 1, electrolyte tank (2), peristaltic pump (3) (BT100-1L, LongerPump, China). A hydraulic seal (4) was connected to the reservoir (2). The hydrogen source was the hydrogen generator GVCH-36A (5) (NPP Khimelectronics, Russia), and hydrogen was moistened through a Drexel flask filled with distilled water (6). The rate of hydrogen supply to the gas diffusion electrode was 2 l/h, the electrolyte circulation rate through the liquid flow electrode was 100 ml/min. Electrochemical measurements were performed on a P-20X galvanostat/potentiostat (Electrochemical Instruments) – (7). The potentiostat was connected to the cell (1) according to a twoelectrode scheme: the working electrode was a liquid flow electrode, and the counter electrode was a hydrogen gas diffusion electrode. The description of the performed electrochemical measurements is presented in 3.1. Measurement parameters.

A flow electrochemical cell used for measurements (see the diagram in Fig. 2) consisted of titanium end plates (1), Viton sheet plates (2), electrically conductive Graflex foil plates (3) (NPO Unimtek, Russia), nickel current collectors (4), graphite electrode plates (5, 8), Teflon flow frames (some with engraved flow channels 6a and the closing part 6b). The hydrogen gas diffusion electrode (7a) was

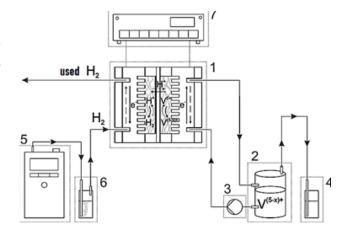


Fig. 1. Installation for determining the concentration and oxidation degree of vanadium in the electrolyte: 1 – flow electrochemical cell; 2 – reservoir for the analyzed electrolyte solution; 3 – peristaltic pump; 4 – water seal; 5 – hydrogen generator; 6 – hydrogen humidifier; 7 – potentiostat/galvanostat

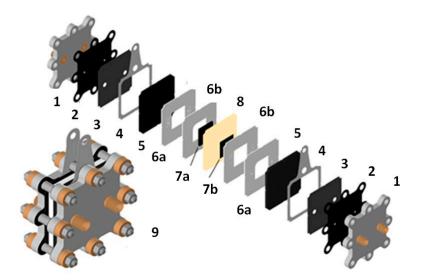


Fig. 2. Scheme of hydrogen-vanadium flow cell used in the experiment: 1 – end plates with fittings for supplying electroactive media (hydrogen and aqueous sulfate solution with vanadium electrolyte sample); 2 – elastomer sealing gasket; 3 – graphite foil gasket; 4 – metal foil terminals; 5 – graphite current-collector plates of the hydrogen electrode; 6 – flow frame of the electrode space area, consisting of two Teflon gaskets (6a, 6b); 7a – hydrogen gas diffusion electrode; 7b – carbon felt electrode for vanadium half-cell; 8 – cation exchange membrane; 9 – scheme of assembled cell

a Freudenberg H23C8 carbon paper coated with Pt/C catalyst (0.516 mg/cm²), and a carbon felt (Sigracell GFD 4.65EA-TA, SGL GmbH, Germany) (7b) was used as a porous flow electrode, while the proton exchange membrane (8) was made of a perfluorinated cation exchange membrane IEM-N102 (Zhongding New Energy Co., Ltd, China).

2.1. Measurement method for determining the concentration and average degree of oxidation of vanadium ions in a vanadium electrolyte using coulometric analysis

To analyze the sample, we used the apparatus presented in Fig. 1. For testing, we assembled a membrane electrode block (Fig. 2). The MEA was connected to the potentiostat using a two-electrode scheme so that the working electrode was made of a carbon felt, and a hydrogen gas diffusion electrode with a catalytic layer of Pt/C was the counter electrode.

The hydrogen was to be supplied to the gas diffusion electrode. To do this, we turned on a hydrogen generator (5) (GVCh-36A (NPP Khimelectronika, Russia)) and a peristaltic pump which circulated hydrogen through a moistening tank (6). Thus, a constant excess

pressure of hydrogen ~20 mbar was maintained in the tank, regardless of whether hydrogen was consumed or generated on the cell during the measurements.

The carbon felt had to be moistened on the cathode side with a background electrolyte of $4\,\mathrm{M}$ $\mathrm{H_2SO_4}$. To do this, we poured 20 ml of background electrolyte into the reservoir (2) and switched on the peristaltic pump (3). The pumping procedure of the background electrolyte for the newly assembled membrane electrode unit lasted about 30 minutes. The rate of hydrogen pumping through the gas diffusion electrode was $2\,\mathrm{l/h}$, and the rate of electrolyte pumping on carbon felt was $100\,\mathrm{ml/min}$.

If the unit is used again, the carbon felt must be washed with a background electrolyte solution before analysis. To do this, we turned on the peristaltic pump (3), poured 20 ml of background electrolyte into the tank (2), waited 5-10 minutes, and drained this electrolyte from the tank. This procedure must be repeated two more times.

To determine the Q_i^0 charges, a sample of the background electrolyte was introduced into the reservoir (2), and the electrochemical measurement procedure was conducted in 4 stages, described in 3.1. Measurement parameters.

The background electrolyte was drained after the procedure.

For the analysis itself, we added 10 ml of the background electrolyte and 1 ml of the test sample to the tank (2). Vanadium electrolyte tends to remain on the walls of the pipette, which was used to take the 1 ml of the sample. Therefore, to reduce the error in the analysis, it is recommended to introduce the test sample using a pipette, then, after mixing the sample with a solution of background electrolyte, take a dilute sample solution into this pipette, and drain it back into the tank. This must be repeated 3 times. Thus the remaining vanadium electrolyte can be washed from the walls of the pipette. After that, we performed the electrochemical measurement procedure described in 3.1. Measurement parameters for determining the values of Q charges.

1.2. Methodology of coulometric analysis

The proposed method is based on transferring an electrolyte sample with vanadium compounds that have an unknown total concentration of $C_{\rm v}$ and an average oxidation state of OS to a completely oxidized state at an oxidation state of +5 and then to oxidation state of +4. The balance of charges that are spent on the oxidation and reduction stages are presented in Fig. 3. The charge at the oxidation stage, $Q_{\rm ox}$, is related to the volume of the electrolyte sample, V, and the unknown average degree of oxidation of vanadium in it, OS, by the following ratio:

$$Q_{OX} = (5 - OS) \cdot C_V \cdot F \cdot V, \qquad (1)$$

where C_v is the concentration of all vanadium forms in the sample and F is Faraday constant. The charge passed through at the recovery stage, $Q_{\rm RED}$, can be calculated using the formula:

$$Q_{RED} = C_V \cdot F \cdot V . \tag{2}$$

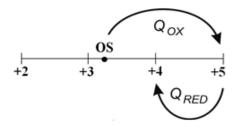


Fig. 3. Scheme of coulometric determination of the oxidation state and concentration of vanadium

Therefore, both unknowns, C_v and OS, were determined from the solution of the system of equations (1), (2):

$$OS = 5 - Q_{RED} / Q_{OX}, \qquad (3)$$

$$C_V = Q_{RED} / (F \cdot V). \tag{4}$$

Thus, the task of finding $C_{\rm V}$ and OS is reduced to measuring $Q_{\rm OX}$ and $Q_{\rm RED}$. And to do this, the conditions must be chosen that ensure the uniqueness of the electrochemical reactions of the conversion of vanadium redox forms, as well as to determine the conditions for conducting and completing the oxidation and reduction stages. This requires using a suitable coulometric cell, the development of a measurement procedure, and testing it on model compositions with known $C_{\rm V}$ and OS. The following part is devoted to this issue.

3. Results and discussion

3.1. Measurement parameters

In a hydrogen-vanadium cell with a membrane-electrode block of the composition $H_2(Pt-C)|IEM|V^{(n-1)+}/V^{n+}$, where n varies from 3 to 5, the following reactions occurred during the measurements:

$$V^{(OS)} \to V^{(+5)} + (5 - OS)e^{-},$$
 (5)

$$V^{(+5)} + e^- \to V^{(+4)},$$
 (6)

$$H^+ + e^- \leftrightarrow 1/2H_2, \tag{7}$$

which means that the release/oxidation of hydrogen on a gas diffusion electrode was used in order to change the degree of oxidation of vanadium. Monitoring the potential of a liquid flow electrode was necessary to determine the completeness of the reactions (5, 6). In turn, the potential of a vanadium flow electrode can be controlled at a given cell voltage on condition that the hydrogen electrode is slightly polarizable. This was facilitated by the high exchange current of the reaction (7) as compared to both reactions (5, 6). To ensure the proper fulfilment of the condition of low polarizability of the gas diffusion electrode, hydrogen moistened at room temperature was supplied to it during the entire measurement, and the vanadium electrolyte sample was additionally diluted before measurement with an acidic background electrolyte in a ratio of 1:10. Such dilution not only reduced the polarization of the hydrogen electrode due to a decrease in flowing currents but also significantly decreased the number of vanadium ion transfers due to an increase in the ratio of concentrations of H+ and vanadium cations.

To select the cell voltages corresponding to the complete oxidation of vanadium to the oxidation state of +5 and its reduction to the oxidation state of +4, and at the same time excluding side reactions (oxidation of electrolyte components or electrode material in the first case and deeper reduction of vanadium in the second), voltammograms were recorded when a background electrolyte of 4 M H₂SO₄ and a vanadium electrolyte with an oxidation state of +4 were supplied to the liquid flow electrode (Fig. 4a, b).

The low current section on the background curve in Fig.4 (a) up to 1.3 V allowed assuming that at voltages less than 1.3 V, the current flowing through a cell with a vanadium electrolyte would be mainly determined by the conversion processes of vanadium compounds (i.e., the background current would be minimal). Judging by the curve in Fig. 5b, the undesirable process of vanadium reduction (+4) with a noticeable rate would begin at a voltage of less than 0.4 V. Thus, to ensure maximum current output of the first reaction (4), $U_{\rm max} = 1.3$ V should not be exceeded, and excessive reduction of vanadium compounds to an oxidation state below +4 could be avoided if the second stage

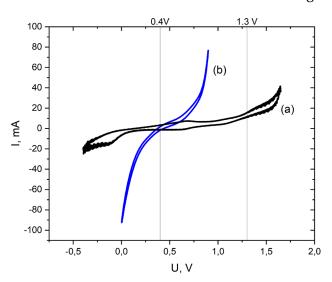


Fig. 4. Voltammograms of the hydrogen-vanadium cell measured at scan rate 20 mV/s for (a) 4 M sulfuric acid solution and (b) model electrolyte with vanadium oxidation state +4 supplied to the liquid electrode, respectively

of coulometry was conducted at voltages not lower than U_{\min} = 0.4 V.

To ensure the shortest duration of the sample analysis, the oxidation and reduction stages should be performed in a potentiostatic mode, setting the $U_{\rm max}$ and $U_{\rm min}$ measured above at the appropriate stages. However, the imposition of such great voltage surges will be accompanied by high pulse currents. The magnitude of these currents can be evaluated based on the data on the cell resistivity obtained by the electrochemical impedance method. This value calculated by the high-frequency cutoff of the hodograph, was about 0.4 Ohm×cm², which gave the pulse current density at the moment of switching U_{max} to U_{\min} : $(U_{\max} - U_{\min})/0.4$ Ohm×cm² = 2.25 A/cm². The working surface area of the MEA (Fig. 2) was 4 cm², so the pulse current density for such MEA was $\sim 9 \text{ A/cm}^2$.

As the total missed charge had to be accurately registered, measuring such currents (great at the beginning of each stage and close to zero at the end) significantly increased the requirements for the potentiostat in the installation. To get around this problem, each of the oxidation and reduction stages was divided into two steps: galvanostatic,

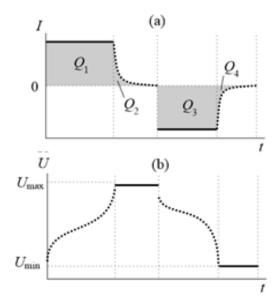


Fig. 5. Schematic dependencies of current (a) and voltage (b) on time during the two-step procedure of complete oxidation and partial reduction of vanadium compounds with each stage divided into galvanostatic and potentiostatic steps. Solid lines indicate the characteristics imposed on the cell, while dashed lines represent the measured values

on condition that it is completed upon reaching the limit voltage, and potentiostatic, that is maintaining the limit voltage until a small stationary current of the appropriate direction is reached. This scheme of two-stage cell polarization with the division of each stage into 2 steps is presented in Fig. 5, which shows separately the dependence of current (a) and voltage (b) on time. The Q_i charges measured at each of the four steps (where i is the step number) were related to $Q_{\rm RED}$ and $Q_{\rm OX}$ as follows: $Q_{\rm OX} = Q_1 + Q_2$, a $Q_{\rm RED} = Q_3 + Q_4$. It should also be taken into account that when

switching the current direction, the double electric layer (DES) was recharged on electrodes with a developed surface, as well as the electrochemical conversion of electroactive compounds that were not included in the aliquot of the analyzed electrolyte (the impurities contained in the background electrolyte, MEA materials and flow circuits of the installation, mainly the results of the analysis of the previous electrolyte portion with multiple repetition of the measurement). To reduce the errors caused by the above-mentioned factors, a preliminary measurement stage was introduced, which consisted of using the same polarization scheme of the cell (Fig. 5) with only the background electrolyte circulating in its liquid circuit. As a result, the values of Q_i^0 were determined. These values were used to correct the Q_i charges measured during the analysis of the vanadium electrolyte. Thus, the final formulas for calculating the average degree of oxidation of OS and the concentration of C_v of a vanadium electrolyte sample were as follows:

$$OS = 5 - \frac{\left(Q_3 - Q_3^0\right) + \left(Q_4 - Q_4^0\right)}{\left(Q_1 - Q_1^0\right) + \left(Q_2 - Q_2^0\right)},$$
(8)

$$c_V = \frac{\left(Q_3 - Q_3^0\right) + \left(Q_4 - Q_4^0\right)}{FV}.$$
 (9)

Ratios (8) and (9) include the absolute values of the Q_i and Q_i^0 charges.

Figure 6 shows the results of applying this sequence of measurements to sample 1. As a condition for the termination of potentiostatic steps 2 and 4 we chose a decrease in the absolute current value to 3 mA/cm². This value was established experimentally as it was assumed by a stationary current that compensated for a change in the degree of oxidation of vanadium compounds as a result of crossover to the anode. Since this value was determined by the transport characteristics of the membrane and the catalytic layer of the anode with respect to vanadium compounds, its clarification was required for each new assembly of the MEA with materials of a different brand.

Table 1 shows the values of Q_i charges according to Fig. 6, as well as the values of Q_i^0 at the preparatory stage of the background electrolyte analysis.

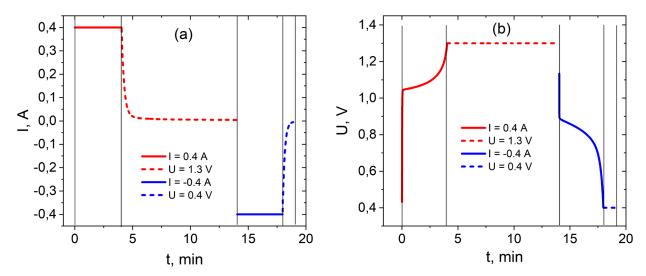


Fig. 6. Dependencies of current (a) and voltage (b) on time during the analysis of sample 1. Solid lines represent galvanostatic segments, while dashed lines represent potentiostatic segments. The legend denotes the parameters set at each of the 4 steps

Table 1. Charge values obtained during the analysis of sample 1 and the background electrolyte 4 M H₂SO₄

Procedure	Sample 1		
step number (i)	Q_i^0 , C	Q_i , C	
1	0.13	78.71	
2	0.05	24.28	
3	0.27	99.56	
4	0.03	2.56	

Based on the charge values indicated in Table 1, we calculated the concentration of vanadium ions C_v using formula (9), it was 1.05 M, and the average oxidation state of vanadium *OS* was calculated using formula (8), it was 3.99.

3.2. Reproducibility of research results

To assess the reproducibility of the method, we conducted a series of measurements on sample 1 and on sample 2. The results of these measurements are presented in tables 2 and 3, respectively.

To increase the accuracy of measurements of concentrations and degrees of oxidation, all experiments were performed on a single MEA. There was no increase in the deviation of the average concentration for sample 1 and sample 2 with the experiment number. It was random, which is an error in aliquot selection. The deviation of the values of the average degree of oxidation in all experiments for sample 1 and sample 2 was less than 1%.

Also, during the experiments we remeasured the correction values of charges on the background electrolyte $Q_{(i)}^{\ 0}$ each time, while an increase in $Q_{(i)}^{\ 0}$

Table 2. Comparison of measurement results of the average oxidation state and vanadium concentration in sample 1 resulting from 4 experiments

. 1 . 27	Sample 1			
Analysis No	$C_{\rm v}$, M	OD		
1	1.05	3.990		
2	1.02	3.993		
3	1.07	3.994		
4	1.01	3.990		
Average	1.04	3.992		
	0.014	0.001		
Confidence interval	1.04±0.04	3.992±0.003		

charges, which could indicate contamination and/or degradation of MEA materials, was not observed. Therefore, the developed cell is expected to have a fairly large resource.

3.3. Analysis of mixed acid electrolyte

To assess the performance of the proposed method for measuring the average degree of oxidation and concentration of vanadium electrolyte for other compositions of vanadium electrolytes in relation to mineral acids, we selected the mixed acid vanadium electrolyte, which is used in commercial vanadium flow batteries (sample 3).

To study the mixed acid electrolyte sample, we assembled a new cell where electrode materials and a membrane were replaced. The background electrolyte used to dilute the 1:10 sample was prepared according to the anionic composition of the studied mixed acid electrolyte.

Three experiments were conducted to determine the average degree of oxidation and concentration of the mixed acid vanadium electrolyte. Table 4 shows the values of the average degree of oxidation and concentration of vanadium electrolyte calculated using the measurement results for sample 3.

Table 3. Results of determining $C_{\rm v}$ and OS for sample 2

A 1 . NY	Sample 2			
Analysis No	$C_{\rm v}$, M	OD		
1	1.610	3.51		
2	1.597	3.49		
3	1.602	3.49		
4	1.612	3.49		
Average	1.605	3.495		
Standart deviation	0.004	0.005		
Confidence interval	1.605±0.011	3.495±0.016		

Table 4. Results of determining $C_{\rm v}$ and OS for sample 3

4 1 . 37	Sample 3		
Analysis No	$C_{\rm v}$, M	OD	
1	1.440	3.38	
2	1.460	3.39	
3	1.480	3.39	
Average	1.460	3.39	
Standart deviation	0.011	0.003	
Confidence interval	1.46±0.04	3.387±0.011	

The average concentration of vanadium ions in a mixed acid electrolyte, calculated using the proposed method, was 1.46 M, which, according to the method, was within the range of permissible vanadium concentrations.

Although the spread of the results in the determined vanadium concentration for all three samples did not exceed 3% of the average value of the parameter being determined, it still significantly exceeded the spread of the OS measurement results (less than 0.3%). This parameter was determined with higher accuracy and reproducibility, as only coulometric data were used for its calculation, while the error in determining $C_{\rm V}$ included an error in aliquot selection.

4. Conclusions

We proposed a method for determining the main characteristics of the VFB electrolyte, which allowed obtaining values of the average degree of oxidation OS and concentration of vanadium ions C_{v} with a relative deviation from the average at 3% for concentration and 0.3% for degree of oxidation, with a duration of one measurement from several minutes to several dozens of minutes. At the same time, it required a laboratory with a fairly simple and inexpensive equipment: a hydrogen source, a peristaltic pump with a capacity of 20-100 ml/min, and a potentiostat operating with currents in the range from 1 mA to 0.5 A, as well as a measuring cell with a MEA area of several cm2. This method will become popular at the sites of production of the VFB electrolyte, as well as in research laboratories dedicated to the development of the VFB area: the devices, installations, and materials necessary for its implementation are included in the basic equipment or are easily accessible, the measurement technique and result processing are simple, and the analytical characteristics are similar to the spectrophotometric method most commonly used for electrolyte analysis. The principle of the method, its experimental design, and the developed procedures (with slight variations) can be generalized to current-forming reactions with other electroactive substances in aqueous electrolytes, as they are based on their basic characteristic - the ability to conduct a redox transformation on electrodes with an approximate current output of 100% over a known potential range.

Author contributions

D. V. Konev – research concept, methodology development, scientific supervision. E. A. Petukhova – conducting research, text writing. E. A. Ruban, V. S. Ershova – conducting experiments. A.V. Terentyev – review writing. R. D. Pichugov – scientific editing of the text. A. A. Usenko – project administration.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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The role of nanofillers of various nature in the morphological changes of the polymer binder for plywood production

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Abstract

Nanomodification of the adhesive composition is one of the promising methods for improving the quality of wood laminated materials. Morphological changes in nanostructures make it possible to control the functional characteristics of the resulting nanocomposites. The shape and distribution of nanomodifiers of nanocellulose and multi-walled carbon nanotubes of urea-formaldehyde resin used to produce plywood have been studied by atomic force microscopy. The phase composition and crystal structure of biological and carbon nanofillers of the binder are investigated. Data on the qualitative and quantitative composition, structural state of nanocrystalline cellulose and multi-walled carbon nanotubes, as well as cured resin in pure and modified form, were obtained by X-ray diffractometry. The microrelief of the surface of the cured binder in the presence of multi-walled carbon nanotubes is characterized by uniformly distributed nanoclusions of 50 nm - 1 µm; and nanoclusions of 70 nm - 2 µm in the case of resin modification with nanocrystalline cellulose. Unmodified ureaformaldehyde resin is characterized by a low degree of crystallinity: the crystallite size is 10 nm. When modifying the resin with multi-walled carbon nanotubes, the crystal size increases to 18 nm, and when modified with nanocrystalline cellulose - up to 15 nm. The most probable type of lattice of the resin under study is a primitive cubic one with a parameter a = 0.840nm. An increase in the volume of the unit cell of resin modified with multi-walled carbon nanotubes (a = 0.844 nm) and nanocrystalline cellulose (a = 0.842 nm) is observed. An increase in the size of the crystalline regions in the resin, as well as an increase in the volume of the resin unit cell as a result of the use of nanomodifiers, can help improve the performance of plywood.

Keywords: Crystal lattice, Urea-formaldehyde resin, Multi-walled carbon nanotubes, Nanocrystalline cellulose, Modification *Acknowledgments*: Authors thank the reviewers for their contribution to the peer review.

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1. Introduction

Russia is one of the leaders among plywood producing and exporting countries [1]. The need for plywood production is due to its wide range of applications in various fields of production [2]. A relevant direction in plywood production is the use of nanotechnology to improve the quality of laminated materials [3, 4]. Unlike other thermosetting resins, urea-formaldehyde resin has a crystalline structure, which makes it possible to strengthen the structure of the polymer matrix and improve its mechanical characteristics [5]. The crystalline regions are favorable for the hydrolytic stability of plywood [6]. Ureaformaldehyde (UF) resin nanomodification is able to influence the size and dispersion of crystalline regions in the resin and comprehensively improve the performance characteristics of the material [5]. It is known that the composition of CF resin affects the degree of crystallinity and crystal morphology in the binder structure [6]. The crystallinity of the resin is preserved during the curing of the CFS in contact with the wood veneer, but the order of the crystal lattice is disrupted [7]. Bio- and carbon nanomodifiers of carbamideformaldehyde binder for plywood production were selected for the study: nanocellulose and multi-walled carbon nanotubes. Morphological changes of nanostructures allow controlling their functional characteristics.

Thus, the purpose of this work is to analyze by atomic-force microscopy (AFM) and X-ray diffractometry (XRD) the morphology of the nanostructure of urea-formaldehyde (UF) resin when nanocellulose (NCC) and multi-walled carbon nanotubes (MWCN) are introduced into it.

2. Objects and methods

The object of the study is a binder based on KF-Zh-F grade UF resin, modified with multi-walled carbon nanotubes and nanocrystalline cellulose.

For the experiment, carbamide-formaldehyde resin of the KF-Zh-F brand (Uralximplast PJSC, N. Tagil, Russia) was used as a polymer matrix. The nanofillers were MWCN with an average diameter of 9.5 nm and a length of 1.5 µm (Nanocyl Technology, Sambreville, Belgium) and NCC with an average particle size (width x length): 10–20 × 300–900 nm (Nanografi Nano Technology,

Ankara, Turkey). The active substrate was peeled birch veneer (*Betula pendula* L.), produced by GalichLes LLC (Galich, Russia), and the passive substrate was glass. Ammonium chloride was used as a resin hardener (Komponent-Reagent LLC, Moscow, Russia).

A suspension of MWCN in an aqueous solution of lauryl sulfate, a surfactant, was obtained by ultrasonic dispersion under constant cooling for 30 minutes on an ultrasonic dispersant UZD-0.1/22 with a power of 100 W and a frequency of 22 kHz. The concentration of the dispersed phase in the suspension was 4.5%.

To obtain the UF resin/MWCN adhesive composition, the components were mechanically mixed for 10 minutes with the addition of MWCN at a concentration of 1.25 wt. %. Ultrasonic dispersion of the adhesive composition in the resin–filler system was carried out under constant cooling for 3 minutes on an ultrasonic dispersant UZK-1.3 with a power of 80 W and a frequency of 24 kHz. Cooling was carried out in a water bath to a temperature of 20 °C in order to avoid polymerization of the adhesive composition. The hardener was added in an amount of 1 wt. %. The adhesive composition of UF resin /NCC was obtained similarly by adding NCC with a concentration of 2 wt. %.

Then the samples of the modified and unmodified resin were cured in a drying cabinet chamber on an inert glass substrate and an active veneer substrate at a temperature of $115\,^{\circ}$ C, after which they were kept for 24 hours at room temperature and humidity of $65\pm5\%$ to complete the polymerization process.

The shape and distribution of nanofillers in the resin were studied using an NT-MDT scanning probe microscope in topology and phase contrast modes. Images of the surface of an unmodified binder cured on an active substrate (veneer), as well as a binder modified with MWCN or NCC, were obtained.

The phase composition and crystal structure of the initial MWCN and NCC powders were studied by electron diffraction in the "lumen" mode on an E'G-100M electronograph. Sample preparation was carried out by ultrasonic dispersion of the test material in surfactants, followed by application to a mesh and drying at a temperature of 20 °C until complete drying.

The phase (qualitative and quantitative) composition and structural state of the nanofillers and the cured resin in its pure and modified form were determined by X-ray diffractometry using a Burken D2-Phaser X-ray diffractometer (Germany). X-ray images of the samples were obtained using CuKa radiation (λ =1.5406 Å) in the angle range 20 from 7 to 70°. During the survey, the samples were placed on a silicon wafer with a "zero" background. To determine the lattice parameters of the crystalline phases and estimate the average size of the crystallites, the profiles of X-ray diffractograms were analyzed using the Rietveld method in the HighScore Plus program.

3. Results and discussions

Atomic-force microscopy in the topology and phase contrast modes has been used to obtain images of the surface of a binder cured on an active substrate (veneer), modified with MWCN

(Fig. 1) or NCC (Fig. 2).

The microrelief of the surface of the resin modified with MWCN is represented by inhomogeneities about 1 µm in size (Fig. 1a). The AFM scan obtained in the topology mode (Fig. 1b) shows nanoscale inhomogeneities in the form of protrusions with a lateral size of about 50 nm. The image obtained in phase contrast mode (Fig. 1c) shows rounded irregularities up to 70 nm in size, darker than the background (indicated by arrows in Fig. 1c). The absence of corresponding inhomogeneities in the image obtained in the surface topology mode indicates that the dark inhomogeneities are nanoinclusions with different mechanical properties compared to the polymer matrix. The nanoscale inhomogeneities very likely correspond to the location of the MWCN.

The microrelief of the resin surface modified by NCC (Fig. 2a) is represented by large,

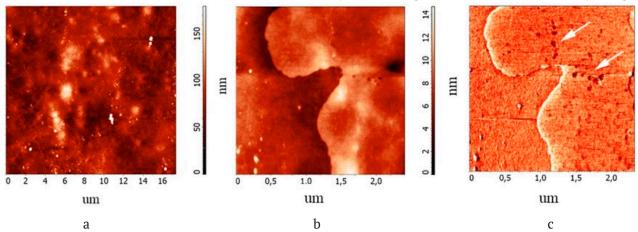


Fig. 1. Images of the surface of a binder modified by MWCN cured on an active substrate (veneer)

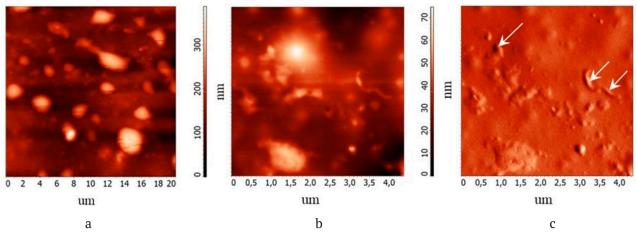


Fig. 2. Images of the surface of a binder modified by NCC cured on an active substrate (veneer)

evenly distributed inhomogeneities with a lateral size of 0.5-2 µm and a height of up to 0.5 µm. The microrelief is poorly developed between the inhomogeneities. The relief of the second level in the areas between large inhomogeneities is represented by irregularly shaped inhomogeneities measuring 70–300 nm (indicated by arrows in Fig. 2c), obviously corresponding to inclusions in the form of fibers (Fig. 2b, c). There is no pronounced phase contrast from the inclusions, which can be explained by the similar elastic-plastic characteristics of NCC and resin.

Figure 3 shows the electronograms of the initial components: powders of MWCN (a) and NCC (b).

The electronogram (Fig. 3a) shows a complete set of reflections corresponding to the MWCN [8]. In the electron diagram (Fig. 3b), all the observed reflections correspond to the crystal lattice of cellulose [9].

Figure 4 shows X-ray diffractograms (XRD) of NCC powder (Fig. 4, curve 1); MWCN powder after dispersion in surfactants (sodium lauryl sulfate (SLS)) and subsequent drying (Fig. 4, curve

2); mixtures of NCC and MWCN after dispersion in surfactants and subsequent drying (Fig. 4, curve 3).

On the XRD of the NCC powder (Fig. 4, Curve 1), very blurred maxima are observed, the position of which coincides with the reflection spectrum from the cellulose crystal lattice. The width of the maxima indicates the nanocrystalline structure of the cellulose.

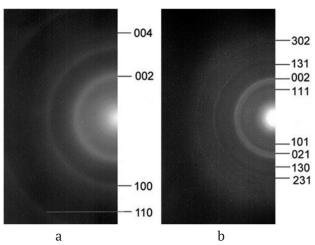


Fig. 3. Electronograms of MWCN (a) and NCC (b) powders

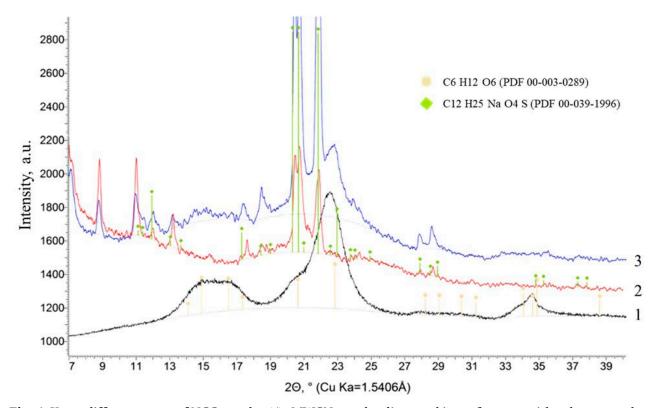


Fig. 4. X-ray diffractograms of NCC powder (1); MWCN powder dispersed in surfactants with subsequent drying (2); mixtures of NCC and MWCN dispersed in surfactants with subsequent drying (3)

All reflections corresponding to the crystalline SLS are observed on the MWCN + SLS XRD (Fig. 4, Curve 2). The absence of reflections corresponding to MWCN is due to both the low-dimensional nature of the nanotubes and their relatively small number.

All the reflections present in the first two diffractograms are observed on the XRD from MWCN +NCC+ SLS (Fig. 4, Curve 3). Reflections associated with phases other than NCC and LSN are not observed, which indicates the absence of crystalline products of the reaction of MNT with NCC in the sample.

Figure 5 shows X-ray diffractograms (XRD) of an unmodified binder (Fig. 5, curve 1) and a binder modified by NCC (Fig. 5, curve 2) and MWCN (Fig. 5, curve 3).

The XRD of unmodified resin (Fig. 5, Curve 1) is represented by wide maxima in the form of halos, the most intense of which is in the angle range $2\theta = 18-28$ °. A low-intensity peak with a peak of $2\theta = 21.11$ ° is observed against the background of this halo. The character of the maxima observed on the diffractogram indicates a low degree of crystallinity of the resin.

In the composite binder containing MWCN

(Fig. 5, curve *3*), in addition to the halo, there are 2 pronounced maxima with a peak of $2\theta = 21.03^{\circ}$ and 23.6° . The higher intensity of the first maximum compared to the XRD of the resin without filler and the appearance of the second maximum indicate an increase in the degree of crystallinity of the resin with filler from MWCN (an increase in the size and proportion of crystallites). The reflections characteristic of MWCN are not observed on XRD, probably due to their small volume fraction in the binder.

Diffraction patterns similar to those of unmodified and modified CNTs of ureaformaldehyde resin are described in the work for formaldehyde resin with different molar ratios. [10]. The authors explain the appearance of pronounced peaks on the resin diffractograms by increasing the degree of polymer crystallinity due to the formation of hydrogen bonds between linear molecules [10].

A halo similar to the unmodified resin is observed on the RD of the resin modified by the NCC (Fig. 5, Curve 2). There are also peaks with peaks of $2\theta = 21.08^{\circ}$ and 34.8° . According to [11], these maxima correspond to reflections from the cellulose crystal lattice.

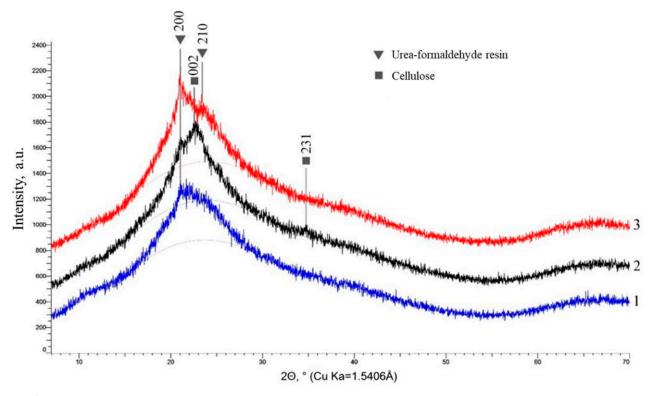


Fig. 5. X-ray diffractograms of unmodified binder (1) and binder modified by NCC (2) and MWCN (3)

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The size of the crystallites (coherent scattering regions), calculated as a result of profile analysis of X-ray diffractograms in the region of the maximum with a peak of 21.1°, was about 10 nm for resin without filler, 15 nm for resin with nanocellulose, 18 nm for resin with MWCN. The effect of increasing the size of crystallites on the strength characteristics of the resulting nanocomposite plywood is confirmed by experimental data. Thus, the tensile strength when chipping through the adhesive layer of plywood modified with MWCN increases by 76% (from 0.68 MPa to 1.2 MPa), modified with NCC – by 194% (from 0.68 MPa to 2 MPa) compared with unmodified plywood [12].

Thus, based on comparative diffraction analysis of unmodified and modified resin samples, the following can be noted: a resin containing ultrasound-activated MWCN [12], after curing, has a higher degree of crystallinity compared with unmodified resin. Modification of the NCC resin activated by ultrasound [13] leads to a less pronounced increase in the degree of its crystallinity.

The most likely type of resin lattice is a primitive cubic one. The cubic lattice constant a was 0.840, 0.844, and 0.842 nm for pure resin, UF resin/ MWCN, and UF resin /NCC, respectively. There is a weak tendency to increase the volume of the unit cell when using MWCN as a nanomodifier of KF–Zh-F resin. For NCC resin, the indices of the observed reflections are 200 and 210. Such a set of reflections may indicate a virtually one-dimensional arrangement of polymer fibers parallel to each other with a relatively arbitrary radial rotation. The reflection indices 002 and 231 are characteristic of NCC powder, which is confirmed by the electronogram data (Fig. 3b).

4. Conclusion

The microrelief of the surface of the modified UF resin/MWCN binder cured on an active substrate is characterized by the presence of uniformly distributed MWCN nanoclusions measuring 50 nm – 1 μ m. The microrelief of the resin surface modified by NCC shows the presence of evenly distributed nanocellulose inclusions measuring 70 nm – 2 μ m. The sets of reflections observed on the electronograms of the nanomodifiers correspond to the MWCN and NCC.

The data of X-ray diffractometry of unmodified resin indicate a low degree of its crystallinity. The modification of the binder of both MWCN and NCC is accompanied by an increase in the degree of crystallinity of the polymer. The crystallite size of the resin without filler is 10 nm; the resin modified with nanocellulose is 15 nm; the resin nanomodified with MWCN is 18 nm. An increase in the size of the crystallites in the resin can lead to an increase in its density, strength, hardness of the material and chemical resistance.

The most probable type of UF resin lattice is a primitive cubic one with parameter a = 0.840 nm. The cubic lattice constant of the resin with MWCN and NCC is 0.844 and 0.842 nm, respectively, i.e. there is a weakly pronounced increase in the volume of the unit cell of the UF resin modified with MWCN.

Thus, the atomic force microscopy data allow us to conclude that the reinforcing components (MWCN, NCC) are evenly distributed in the polymer matrix. The AFM data on the morphology of the surface of composites with NCC obviously correlate with the XRD data: according to the first method, morphological signs of the presence of fibers are observed, according to the second, signs of the NCC crystal lattice. For MWCN, AFM morphology data correlate with electronography data. Electronography confirms that the reinforcing material has a crystalline structure of MWCN. And AFM reveals the presence of inclusions with other elastic-plastic properties in the polymer. Taken together, both methods indicate that the inclusions in the matrix are precisely the MWCN. X-ray diffractometry proved to be insensitive in the case of MWCN.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Theoretical and experimental study of the niobium dioxide electronic structure

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Abstract

The investigation of the niobium dioxide electron-energy structure is presented in the paper. The electronic structure computer modeling of the NbO_2 with a rutile crystal structure has been performed using linearized augmented plane wave method. The energy band structure as well as total and partial densities of electronic states are calculated.

Experimental studies of the NbO₂ sample electronic structure were carried out using synchrotron and laboratory X-rays sources. The X-ray photoelectron spectrum of the valence band and subvalent states of NbO₂ and the spectrum of the X-ray absorption near edge structure near K-edge of the oxygen atom in NbO₂ have been recorded.

The spectra of the X-ray absorption near edge structure of the oxygen and niobium atoms K-edges are modeled. The calculated spectra make it possible to reliably interpret the data from the synchrotron experiment. It is shown that for NbO_2 the spectrum calculated for the ground energy state without using the supercell and core hole modeling method demonstrates high agreement with the experiment.

Keywords: Computer modeling, Niobium dioxide, Electronic structure, Density of states, XANES, XPS, Core hole, Rutile *Funding*: The study was supported by the Russian Science Foundation grant no. 22-72-00145, https://rscf.ru/project/22-72-00145/.

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1. Introduction

Materials in which the metal-semiconductor phase transition is observed can be used to create various electronic, optical and other devices: memory elements, neuromorphic hardware devices, "smart windows", devices for energy generation and storage, etc. [1-4]. Such materials include niobium dioxide NbO, [3, 5, 6]. Above the phase transition temperature (for NbO₂ Tc \approx 808 °C), this material has a crystalline structure of the classical rutile type (P4,/mnm space group) and is a conductor of electric current [7, 8]. Below this temperature, the crystal structure of NbO, changes to a distorted rutile structure (space group I4,/a) that accompanied by a change in the nature of the conductivity to semiconducting [7, 8].

In order to study the features of the electronenergy structure of niobium dioxide in the present work a comprehensive study is carried out including the use of ab initio methods of computer modeling, X-ray photoelectron spectroscopy (XPS) and X-ray absorption near edge structure (XANES) spectroscopy.

2. Calculation technique

The high-temperature phase of niobium dioxide has a rutile crystal structure, belongs to tetragonal syngony, and is characterized by the $P4_2$ /mnm space group [3, 7, 9]. The experimental values of the NbO_2 crystal structure parameters given in Table 1 [10] were used in the calculation, similar to other works [9, 11, 12]. The appearance of the NbO_2 unit cell and the corresponding first Brillouin zone are shown in Fig. 1a, b.

Calculations of the electronic structure were performed using the linearized augmented plane

Table 1. Parameters of the NbO₂ crystal lattice

Space group	$P4_2/mnm$		1
Unit cell parameters a, b, Å	4.8464		
Unit cell parameter c, Å	3.0316		
Atomic position	x/a	y/b	z/c
Nb ₁	0.0	0.0	0.0
Nb ₂	0.5	0.5	0.5
$O_{\scriptscriptstyle 1}$	0.2925	0.2925	0.0
O_2	0.7075	0.7075	0.0
O ₃	0.2075	0.7925	0.5
$O_{\!\scriptscriptstyle{4}}$	0.7925	0.2075	0.5

waves method using the Wien2k software package [13]. The Generalized Gradient Approximation (GGA) was used to account for the exchange-correlation energy [14]. The radii of muffintin spheres $R_{\rm mt}$ given in atomic units of lenght (numerically equal to the Bohr radius $a_{\rm o}$), were 1.98 a.u. for the Nb atom and 1.79 a.u. for the O. The $R_{\rm mt}$ · $K_{\rm max}$ value that determines the number of the basis plane waves was 6.0. When integrating over the Brillouin zone 5000 k-points were used for calculating the unit cell and 200 k-points for calculating the supercell.

The XANES spectra were modeled for a unit cell in the ground energy state as well as for a 2×2×3 supercell using the core hole method. The formalism of this method is described in detail in our previous work [15]. Previously, this method was successfully tested when calculating the XANES spectra of various oxide materials [16, 17].

3. Experimental technique

The experimental part of the work consisted in conducting studies of the commercial NbO₂ powder sample produced by Sigma-Aldrich [18] using XPS and XANES spectroscopy. X-ray phase analysis of the sample showed that it has a crystalline structure of distorted rutile (space group I4,/a).

The experiments were carried out at the NanoPES synchrotron radiation end-station including the ESCA module of the National Research Center "Kurchatov Institute" [19]. The pressure in the analytical chambers of the NanoPES workstation spectrometers was $\sim 10^{-10}$ Torr. Electron energy analyzers Specs Phoibos 250 and Phoibos 150 were used. When using a laboratory monochromatic source, the excitation energy was 1486.6 eV.

We used a standard approach to data calibration based on recording the C 1s signal line of hydrocarbon contamination [20]. To compare and analyze the main features of the spectra, well-known databases were used from which the most relevant data were selected [20, 21].

4. Results and discussion

The band structure of NbO_2 is shown in Fig. 1c. The position of the Fermi level $(E_{\scriptscriptstyle F})$ is selected as the beginning of the energy count. The lower part of the valence band consists of

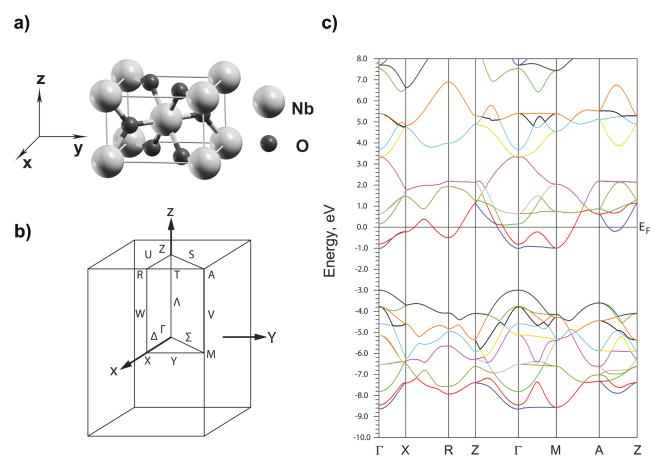


Fig. 1. NbO₂ unit cell with rutile crystalline structure (a). The first Brillouin zone of NbO₂ (b). Band structure of NbO₂ (c)

12 bands originating mainly from 2*p*-states of the oxygen with a small admixture of 4d-states of the niobium (see Fig. 2c, d), has a width of 5.65 eV and is located in the energy range of -8.65.. -3.0 eV. Above the top of this group of bands there is a 1.98 eV wide gap of forbidden energies. Even higher in the energy is a group of 10 bands originating mainly from 4d-states of niobium atoms with a certain proportion of 2p-states of oxygen atoms. According to the crystal field theory due to the octahedral surrounding of the niobium atom by oxygen atoms in NbO, the 4d-states of Nb are splitted into 6 low-energy bands with t_{2g} symmetry and 4 high-energy bands with e_g symmetry [9]. These two groups of 4d bands are well separated in the band structure with the exception of the point Γ vicinity where they slightly overlap [22].

A comparison of the performed calculations results with the data from [8, 9, 11, 22] shows a fairly high agreement which indicates their reliability. At the same time some differences

are observed. Compared with the results of [11] there is a noticeable difference in the behavior of the uppermost $t_{\rm 2g}$ band in the Z-A and Z-R directions connecting the center of the upper face of the Brillouin zone Z with its vertex A and the middle of its edge R respectively. In both cases the bands have an almost flat character but in our calculation this band lies ~ 1 eV lower in these directions (at an energy of ~ 2.2 eV above $E_{\scriptscriptstyle E}$) than in [11]. At the same time our picture of the band dispersion of the uppermost $t_{2\sigma}$ band is closest to the results of work [9]. However, in the calculation [9] as in [8] but unlike [11, 22] there is a gap of forbidden energies separating the upper t_{2g} band from the lower eg band. In our calculation these two groups of bands converge in the vicinity of point Γ similar to the results of [11, 22].

In our calculation the forbidden energy gap dividing the valence band into two parts has a width of 1.98 eV. This is less than the values obtained in other works: 6.1 eV [8], 4.1 eV [22], ~ 3 eV [9].

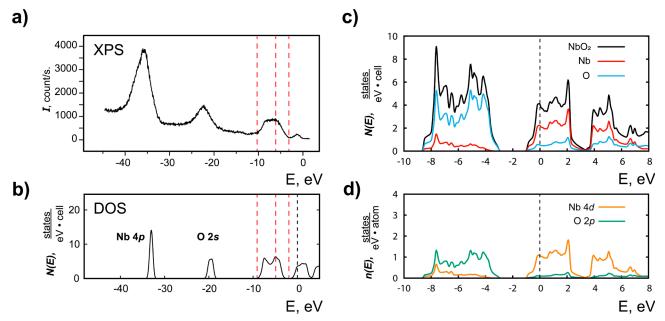


Fig. 2. XPS spectrum of NbO₂ (a). Total DOS of NbO₂ (b). Total DOS of the crystal and the Nb and O atoms in the valence region and near the bottom of the conduction band (c). Partial DOS of the Nb and O atoms (d)

In general, NbO, has a band structure typical of rutile-like transition metal oxides. The most important characteristic defining the unusual properties of NbO₂ is the position of the Fermi level which is explained by the following considerations. The free niobium atom has five electrons on the valence shell occupying $4d^45s^1$ -states. During crystallization into the rutile structure the niobium valence shell electrons are transferring to more electronegative oxygen atoms. Formally the niobium atom in rutile-like NbO, has an oxidation state of Nb⁺⁴ and the filling of the d-subshell is $4d^1$ [23]. As a result, the energy bands formed from the d-states of niobium atoms are partially occupied by electrons. The Fermi level crosses a group of bands formed by Nb 4d-states which acts as a conduction band and NbO₂ exhibits a metallic type of electrical conductivity.

The results of NbO₂ XPS spectroscopy provide important information about the features of occupied electronic states. Fig. 2a shows the X-ray photoelectron spectrum of the NbO₂ sample recorded in the range of binding energies from -45 to 0 eV. In the considered energy range 4 regions of occupied states are observed similar to the results of [24] and the experimental spectrum is generally consistent with the results of [11, 24–28]. The XPS spectrum

was compared with the total density of states calculation results in the same energy range by combining the position of the valence band on the energy scale (Fig. 2b). At the same time the starting point of the experimental and calculated scales do not coincide. Fig. 2a, b shows a good agreement of the results which makes it possible to explain the features observed in the XPS spectrum. The intense peak at -35.8 eV in the experimental spectrum corresponds to the density of Nb 4p- states peak at the -33.0 eV on the calculation scale where the position of the Fermi level is taken as the starting point. The experimental peak at -22.25 eV corresponds to the density of O 2s- states peak at an energy of -19.5 eV on the calculation scale. The peaks position of the Nb 4p- and O 2s- states in the calculation turns out to be shifted closer to the Fermi level if compared with the experiment. At the same time the distance between the peaks in the calculation is 13.5 eV which is in excellent agreement with the experimental value of 13.55 eV.

Features in the energy range -10 ... 0 eV describes the structure of the valence band. This area as well as the bottom of the conduction band are shown on an enlarged scale in Fig. 2c, d. The total density of electronic states (DOS) of the NbO₂ crystal, the total atomic DOS for Nb

and O atoms as well as the partial densities for Nb 4d- and O 2p- electronic states dominating in the considered energy range are shown. As mentioned above the valence band consists of two regions. Its lower part from -8.65 to -3.0 eV is formed mainly by 2p-oxygen states. Nb 4d-states dominate in the upper part of the valence band and at the bottom part of the conduction band in the range from -1 to +6 eV. It can be seen that there is a contribution of O 2p-states to the two upper groups of bands and Nb 4d-states to the lower group respectively that indicates the covalent component of the chemical bond [9]. In the low-energy group of states there are local maxima at the bottom and top of the considered range with the local minimum observed in its center. The Fermi level is located near the local DOS maximum which may indicate the instability of such an electronic structure and contribute to the metalsemiconductor phase transition observed in NbO₂. The calculated DOS spectra are in good agreement with the results of [11, 22].

Figure 3 shows a comparison of the experimentally recorded O K-edge XANES

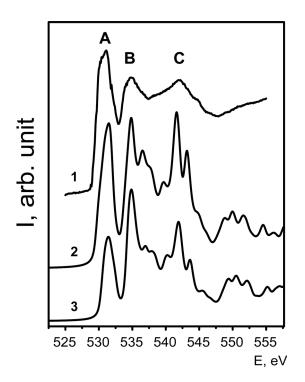


Fig. 3. O K-edge XANES spectrum. *1* – experimental, *2* – calculation for unit cell without core hole, *3* – calculation for supercell with core hole

spectrum with the spectra calculated for a unit cell in the ground energy state (without core hole modeling) and for a 2×2×3 supercell calculated using the core hole method. The experimental spectrum contains three maxima designated as A, B and C at energies 531.1, 534.8, and 542.0 eV, respectively. Maximum A has the highest intensity. The intensities of maxima B and C are approximately equal. Experimental spectra of the O K-edge with the same intensity ratio between the maxima were obtained in [27] by the XANES method for NbO₂ thin films and in [12] by electron energy loss spectroscopy (EELS) for a commercial sample produced by Alfa Aesar. The relative intensity of peak A is less than the intensity of peak B but greater than the intensity of peak C in spectra obtained by EELS for commercial samples produced by Aldrich [29] and Alfa Aesar samples [30]. It can be concluded that the data presented in the literature are ambiguous and probably depend on the conditions of the experiment as well as the subsequent mathematical processing of the result.

The calculated spectra shown in Fig. 3 were compared with the experimental one based on the maximum B position at an energy of 534.8 eV. In our calculation performed for a unit cell in the ground energy state all three maxima have a similar intensity, but maximum A has the lowest value and maximum C has the highest. There are known papers [12, 27] in which calculations of the O K absorption edge were also performed using the Wien2k software package. In [27] the ratio of the maxima A and B have the same intensities as ours but unlike our result in [27] maximum C has the lowest intensity. In [12] maximum A is the most intense followed by least intense maximum B and C. Thus, in all three calculations including the present one the obtained results differ slightly in the ratio of the main spectrum features intensities. At the same time there is a good agreement between the calculated spectra and the experimental one in terms of the number of peaks and their position on the energy scale that makes it possible to unambiguously interpret the experimental data.

The use of the core hole method in the calculation does not lead to a change in the number of features in the spectrum or their position on the energy scale. However, the

relative intensity of the A and C maxima is noticeably decreasing. The decrease of the first peak intensity in the calculated XANES O K-edge spectrum by the core hole method in transition metal oxides is a well-known effect [31]. According to our calculations and ones given in [23, 31] the hybridization of unoccupied O2p-electronic states and unoccupied states of metal atoms is observed in transition metal oxides. According to this approach the O K spectrum fine structure is explained as follows. Peak A in the spectrum reflects the density of oxygen 2p- states hybridized with niobium t_{2g} states, peak B – with niobium t_{2g} states. Peak C reflects the hybridization of to 2p- states and to 2p- states a

intensity associated with t_{2g}-symmetry states [31]. Using the XANES O K spectrum as the example we conclude that the calculation for the NbO₂ unit cell without accounting the core hole gives results that are better consistent with experiment. Therefore, for the Nb K absorption edge the calculation was performed only for the unit cell. The result in comparison with the experimental spectrum [32] is given in Fig. 4. The calculated spectrum reproduces all the absorption edge fine structure main features presented in the experimental spectrum which indicates the correctness of the chosen modeling method and the reliability of the result obtained.

5. Conclusions

The electronic structure of niobium dioxide has been studied by the linearized augmented plane waves method. The band structure, total and partial densities of electronic states are calculated. The DOS spectrum allows reliable interpretation of XPS spectroscopy results.

The results of XANES spectra calculations for the Nb K-, O K- absorption edges are presented. For the O K absorption edge spectrum it is shown that the calculation for the ground energy state without using the supercell and core hole modeling method demonstrates high agreement with the synchrotron experiment.

The results obtained can be used in the analysis of experimentally studied samples of the Nb – O system.

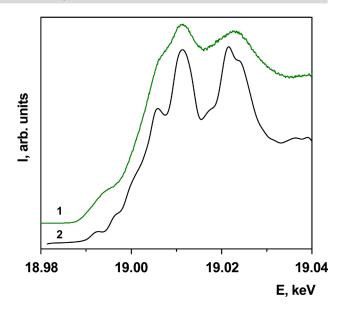


Fig. 4. Nb K-edge XANES spectrum. 1 – experimental [32], 2 – our calculation.

Contribution of the authors

All authors made an equivalent contribution to the preparation of the publication.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have appeared to influence the work reported in this paper.

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Monte Carlo simulation of interfacial adhesion between geopolymer binders and mineral aggregates

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Abstract

Silico-aluminophosphate and alkali-aluminosilicate geopolymers are increasingly popular as a green alternative to traditional Portland cement concrete used in the construction industry. In geopolymer concretes and mortars, the aggregate-matrix interface plays a major role in the fracture mechanisms. The adhesion strength between the mineral aggregate and the geopolymer matrix is mainly determined by the chemical nature of the components of the aggregate-geopolymer interface. However, this aspect remains insufficiently studied. Therefore, we used a Monte Carlo simulation to investigate adhesive behavior and interfacial interaction mechanisms of a cyclic aluminosilicate oligomer forming the structure of a geopolymer gel with mineral aggregates.

The study determined the low-energy equilibrium configurations of the structure of oligomers adsorbed on the surface of quartz, calcite, albite, and microcline, as well as the adsorption energies.

Keywords: Geopolymer, Interfacial adhesion, Mineral aggregate, Quartz, Calcite, Albite, Microcline, Monte Carlo method *Funding:* The study was supported by the Russian Science Foundation grant No. 19-79-10266, https://rscf.ru/project/19-79-10266/.

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☑ Andrey M. Vasilchenko, e-mail: zmandr@mail.ru © Ermolov Y. M., Vasilchenko A. A., Mischinenko V.B., 2025



1. Introduction

Over the past few years, silico-aluminophosphate and alkali-aluminosilicate geopolymers [1] have been considered as a green alternative to traditional Portland cement concrete used in the construction industry [2-4] and as promising binders for a wide range of applications [5, 6]. The adjustable pore microstructure of geopolymers makes them promising materials for thermal, acoustic, and vibration insulation [7–9]. Geopolymers also exhibit effective immobilization properties and can thus be used for the utilization and disposal of radioactive waste [10–12]. Owing to their strong bonding properties, geopolymers can also be used as repair binders for concrete and reinforced concrete structures [13, 14].

In geopolymer concretes and mortars, the aggregate-matrix interface plays a major role in the fracture mechanisms and hence determines the mechanical properties of the aggregate. The properties of the interface affect the process of deformation of geopolymer materials, which typically deteriorate within the interfacial layer. A weak interfacial adhesion between geopolymer binders and mineral aggregates often results in inferior mechanical properties [15]. The adhesion strength between the mineral aggregate and the geopolymer matrix is determined by frictional forces at the aggregate-geopolymer interface and the chemical nature of these components. The impact of the latter factor on the final adhesion strength is more significant. However, this aspect remains insufficiently studied.

In our study, we used a Monte Carlo simulation to investigate the adhesive behavior and the mechanism of interfacial interaction between a cyclic aluminosilicate oligomer forming the structure of a geopolymer gel and characteristic minerals in the structure of an inorganic aggregate (α -SiO $_2$, β -SiO $_2$, calcite, albite, and microcline). The article presents low-energy equilibrium configurations of the aluminosilicate oligomer on the surface of the considered minerals and the results of a comparative analysis of adsorption energies.

2. Experimental

To study the interfacial interaction between the geopolymer obtained by means of lowtemperature alkali activation of aluminosilicates [16] and mineral aggregates, we used a cyclic aluminosilicate oligomer (CAO) with a ratio of Si/Al = 3 shown in Fig. 1a. The structure of the CAO was optimized using the Dmol3 software package [17]. The calculations were performed by means of exchange-correlation functional approximations using the Perdew-Burke-Ernzerhof (RPBE) functional [18]. The electron density was determined using a double numerical plus polarization (DNP) basis set [19]. In order to account for the van der Waals forces, we used a dispersion-corrected DFT (DFT-D) suggested by Grimme [20, 21].

The following surfaces were used for the simulation of the adsorption behavior of the CAO: (100) quartz (α -SiO₂) [22]; (100) quartz (β -SiO₂) [23], (010) calcite (CaCO₃) [24], (100) albite

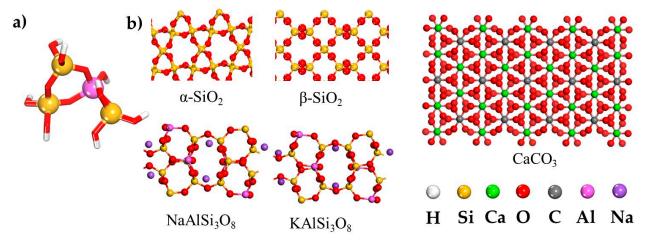


Fig. 1. (a) model of a ring-type aluminosilicate oligomer defining the structure of the N-A-S-H geopolymer gel; (b) surface models: $(100) \alpha$ -SiO₂; $(100) \beta$ -SiO₂; $(010) CaCO_3$; $(100) NaAlSi_3O_8$; $(100) KAlSi_3O_8$

 $(NaAlSi_{z}O_{s})$ [25], and (100) microcline $(KAlSi_{z}O_{s})$ [26] shown in Fig. 1b. In order to eliminated the effect of the atomic layers associated with periodic boundary conditions when simulating a mineral surface, we used a 40 Å vacuum layer. The simulation was performed by means of the Monte Carlo technique in a gas phase using the Adsorption Locator software unit [17]. The calculations were based on the Metropolis-Hastings algorithm and involved simulated annealing with 20 measurement cycles at different temperatures and 2×10^5 steps for each cycle [27]. The energies were calculated using the COMPASS force field [28]. The van der Waals energy was calculated using the atomic method. The cut-off distance of the Lennard-Jones potential was 14 Å. The electrostatic interaction was described using the Ewald summation technique [29].

3. Results and discussion

The interfacial adhesion between the geopolymer binder and the aggregate was assessed based on the results of the Monte Carlo simulation of the adsorption of the cyclic aluminosilicate oligomer on the analyzed surfaces. In our study, we determined the most stable adsorption configurations of the CAO and its adsorption energies.

The adsorption energy $(E_{\rm ads})$ was determined using the formula:

$$E_{\rm ads} = E_{\rm Total} - (E_{\rm Adsorbate} + E_{\rm Surface}),$$

where $E_{
m ads}$ is the energy of the mineral surfaceadsorbate configuration, $E_{
m Adsorbate}$ and $E_{
m Surface}$ are

energies of the CAO and the mineral surface respectively.

The low-energy equilibrium configurations of the structure of the CAO adsorbed on mineral surfaces (α-SiO₂, β-SiO₂, CaCO₃, NaAlSi₃O₈ KAlSi₂O₆) determined as a result of the Monte Carlo simulation are demonstrated in Fig. 2. We can see that as a result of the adsorption, the CAO is oriented parallel to the mineral surface. This spatial conformation of the CAO allows for maximum number of interactions with the mineral surface. To determine the main types of interactions between the adsorbate and the mineral surface, we performed an analysis of bond lengths. The shortest distances between the atoms of the mineral surface and the nearest aluminum and silicon atoms of the adsorbed oligomer in the equilibrium state are given in Table 1. The distances were calculated based on the difference in the Cartesian coordinates. To calculate the number of bonds between the surface oxygen atoms and the hydroxyl groups of the oligomer, we used a geometric criterion, according to which the distance between the atoms forming the hydrogen bond cannot be more than 2.45 Å [30]. In our study, practically all the shortest bonds were less than 2.5 Å, which indicates the formation of a strong bond between chemisorbed oligomers and the mineral surface.

Fig. 3 presents the calculated $E_{\rm ads}$ for all the studied systems. Significantly negative $E_{\rm ads}$ values indicate the strongest and most stable adsorption. Fig. 3 demonstrates that the absolute values of the adsorption energy of the CAO on the studied

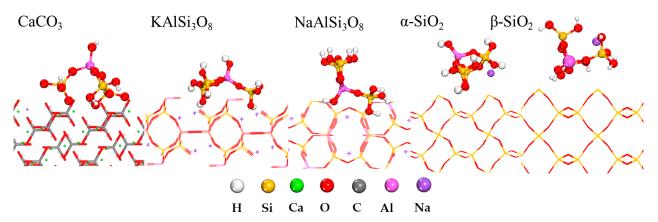


Fig. 2. Most stable low-energy configurations calculated by MC simulations for the adsorption of oligomer on the mineral surfaces of aggregate

Table 1. The distance (Å) between the layer of near-surface atoms and the atoms of the oligomer, as well as the number of bonds formed, calculated by the Monte Carlo method

Mineral aggregate	Al	Si	Si	Si	Number of bonds
CaCO ₃	3.22	5.72	3.36	3.09	5
KAlSi ₃ O ₈	2.78	1.99	2.52	4.83	4
NaAlSi ₃ O ₈	5.07	2.04	5.32	5.67	3
α -SiO $_2$	3.89	1.49	6.61	2.97	2
β-SiO ₂	2.52	4.31	2.05	4.31	2

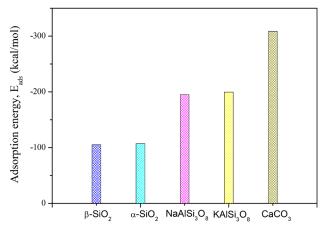


Fig. 3. Adsorption energies obtained from Monte Carlo simulations for oligomer adsorption on the surface of mineral aggregates in the gas phase

mineral surfaces decrease in the following order: $CaCO_3 > KAlSi_3O_8 > NaAlSi_3O_8 > \alpha-SiO_2 > \beta-SiO_2$. This tendency correlates well with the number of formed OH-bonds (Table 1). The calculated E_{ads} demonstrated that the best adhesion is observed during the interaction of the CAO with calcite, which is confirmed by the experimental data presented in [31, 32].

4. Conclusions

This study presents Monte Carlo simulation results investigating the adhesive behavior and the mechanism of interfacial bonding of a cyclic aluminosilicate oligomer and mineral aggregates. The study determined low-energy equilibrium configurations of the structure of oligomers adsorbed on the surface of (100) α -SiO₂, (100) β -SiO₂, (010) CaCO₃, (100) NaAlSi₃O₈, (100) KAlSi₃O₈, as well as the adsorption energies. The simulation demonstrated that cyclic aluminosilicate oligomers can be chemisorbed. The absolute values of the adsorption energy

of the aluminosilicate oligomer on the studied surfaces are characterized by the following sequence: $CaCO_3 > KAlSi_3O_8 > NaAlSi_3O_8 > \alpha - SiO_2 > \beta - SiO_2$. This result is in good agreement with the number of the OH-bonds formed between the aluminosilicate oligomer and the mineral surface. The best adhesion effect was observed for calcite.

Contribution of the authors

The authors contributed equally to this article.

Conflict of interests

The authors declare that they have no known competing financial interests or personal relationships that could have influenced the work reported in this paper.

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Modelling of interdiffusion and phase formation in thin-film two-layer systems of polycrystalline oxides of titanium and cobalt

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N. N. Afonin¹⊠, V.A. Logacheva²

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Funding: The study was funded by the Russian Science Foundation, research project No.

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Patent

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Example:

- **Fig. 1.** Dependences of the parameters a and c of the tetragonal lattice of nanocrystalline PdO films on the oxidation temperature T_{ox} : 1 single-phase PdO films, 2 heterophase PdO + Pd films; 3 data of the ASTM standard [22, 23]
- **Table 1.** The values of relative electronegativity (ENE) of some chemical elements [30] and the proportion of the ionic component of the chemical bond in binary compounds of the AB composition formed by these elements

1100

1200

1000

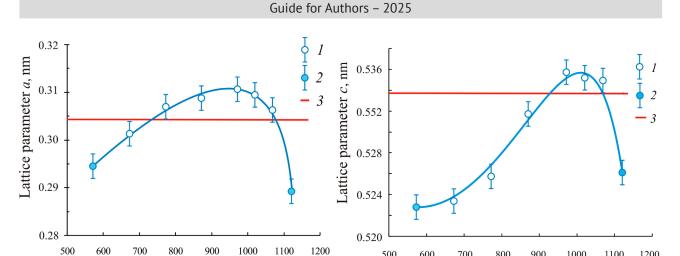


Fig. 1. Dependences of the parameters *a* and *c* of the tetragonal lattice of nanocrystalline PdO films on the oxidation temperature T_{ox} : 1 – single-phase PdO films; 2 – heterophase PdO + Pd films; 3 – data of the ASTM standard [22, 23]

500

600

700

800

Oxidation temperature T_{ox} , K

900

1200

Table 1. Values of the ionic radii of palladium Pd²⁺ and oxygen O²⁻ [30–32]

1100

Ion	Coordination number CN	Coordination polyhedron	Values of ionic radii $R_{ m ion}$, nm
Pd^{2+}	4	Square (rectangular)	0.078 [30]; 0.086 [31]; 0.078 [32]
O ²⁻	4	Tetragonal tetrahedron	0.132 [30]; 0.140 [31]; 0.124* [31]; 0.132 [32]

^{*}The values of ionic radius were obtained on the basis of quantum mechanical calculations.

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500

600

700

800

Oxidation temperature T_{ox} , K

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