



Optimal Monazite Concentration Processes for the Extraction of Uranium and Thorium Fuel Material

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Abstract: The optimal conditions for the nitric acid dissolution of precipitates of hydroxides and hydrated oxides of rare-earth elements, uranium, and thorium obtained after autoclave alkaline opening of samples of monazite concentrate have been determined. The distribution of radioactive impurities between the solid phase and the solution in the processes of alkaline opening, dephosphorization, and acid dissolution of the pulp was studied. Two options are proposed for the extraction of uranium and thorium in the presence of rare earth elements, followed by separation of the components using tributyl phosphate of various contents in the carbon diluent.

Keywords: monazite; autoclave leaching; flow chart; rare earth elements; thorium; uranium; radium; separation; extraction

1. Introduction

The development of nuclear energy programs is presently underway in several countries. In particular, it is included in the list of priority areas of economic modernization for the Russian Federation. The nuclear power industry in the Russian Federation, as in many other countries, was based on the use of the uranium–plutonium fuel cycle, mainly developed to address military problems [1]. However, in the medium-term—probably already within a few decades—the application of the thorium-uranium fuel cycle, based on the use of thorium, for the development of nuclear energy [2] will acquire increasing importance, due to the relative abundance of thorium in nature. Indeed, unlike uranium raw materials, the thorium world reserves (especially in the Russian Federation) are practically unbounded. The Th content of the earth's crust is 8×10^{-4} wt% [3], which is about 3.5 times higher than that of U. About 120 thorium-containing minerals are known [3–7].

Furthermore, the use of thorium provides a number of additional advantages.

Indeed, the extraction of thorium is an essential task in the industry of rare-earth elements (REEs) which are associated with several thorium raw materials [8]. Another considerable advantage when using the thorium-uranium cycle is the elimination of the production of plutonium and trans-plutonium elements, which not only reduces the risk of proliferation of fissile materials, but also reduces the amount of radioactive waste compared with the uranium-plutonium cycle [4]. The introduction and use of the thorium-uranium cycle in nuclear energy can ensure the environmentally safe development of the industry through the use of approaches that exclude the accumulation of long-lived radioactive waste [9].



Additionally, equipment investment and operation costs are considerably lower when using thorium fuel elements because the reactor volume is reduced by a factor of 2–3 for the same amount of energy production and the reactor can be operated continuously for an estimated 50 years without fuel reloading [10].

In view of the undoubted advantages of using the thorium-uranium cycle, it is both important and relevant to search for new, promising sources of thorium raw materials and to create innovative technologies for their processing. The most significant minerals for the industrial extraction of Th are monazite, loparite, and pyrochlore.

The distribution of these minerals in the Russian Federation is briefly reviewed in the next lines. Euxenite sources are located in the territory of the Kola Peninsula [9–12] with a total thorium content of up to 160 thousand tons [13,14], whereas the Tomtorskoye deposits (monazite–pyrochlore) [15] and those of Novotroitskoye, Tarak, Bashelavskoye, Sanarskoye, and Baleyskoye (monazite)—containing up to 5.1-8% of ThO₂ [16]—provide promising opportunities of exploitation. Deposits of tantalite are presently exploited at Etykinsky, Vishnyakovsky, whereas minerals of pyrochlore are extracted at the sites of Tatarsky and Katuginsky [17]. Abandoned sites include Orlovsky (tantalite) and Beloziminsky (pyrochlore) [18]. In total, over 1500 thorium-containing deposits and ore occurrences with reserves of Th up to 1.7 million tons are available in the Russian Federation [10–13].

One of the most important sources of Th production is monazite mineral concentrate presently being stored at the UralMonatsit State Institution near Krasnoufimsk, Sverdlovsk Region. Its chemical formula is Ce,La,Th,Ca[PO₄, SiO₄, SO₄]; it contains from 50% to 68% of REE oxides; from 22% to 31.5% of P₂O₅; from 10% to 28% of ThO₂, including isomorphic ZrO₂ (up to 7%) and SiO₂ (up to 6%) and impurities of MgO, MnO, PbO, Fe₂O₃, Al₂O₃ [19].

The classical technologies for processing monazite, based mainly on the sulfuric or alkaline opening of mineral raw materials followed by extraction of valuable components, are described in [20]. An alternative approach is based on the use of supercritical fluid extraction with carbon dioxide containing tributyl phosphate adducts [21], D2EGFK [22] and other reagents [23–27].

In the present work, new approaches for optimizing the technologies for alkaline processing of monazite [28] are proposed, including grinding of the raw material, acid dissolution of precipitates of REEs, uranium, and thorium oxides-hydroxides, followed by their extraction and separation.

2. Materials and Methods

The experiments that form the basis of this study were conducted employing the monazite concentrate with a bulk density of 2.8 g/cm³ stored at a branch of the UralMonatsit Regional State Institution (the average composition is given in Table 1).

Monazite grinding was carried out using the SM-3 cone crusher (Vilitek LLC, RF, Moscow, Russia). Monazite, disintegrated in the crusher, was dispersed on sieves with multiple returns of a large fraction to the head operation. In the study, a significant amount of monazite concentrates with different grain sizes of <0.05 mm, 0.05–0.125 mm, 0.125–0.25 mm and 0.25–0.5 mm was investigated.

Alkaline opening of the monazite concentrate was carried out in a stainless-steel metal apparatus, and the regimes for transferring the formed pulp to the nitric acid medium were worked out in quartz vessels.

For the opening of raw mineral materials, NaOH of the "chemically pure" qualification was used. Monazite concentrate powder of a certain fraction and known mass was placed in an autoclave, and NaOH solution of a given concentration and volume was introduced.

Literature data confirm [29] that the most important factors affecting the completeness of alkaline opening of monazite concentrate are contact time, temperature, and degree of fineness of the initial material; on the contrary, a change in the alkali content in the solution and the solid-to-liquid ratio affect it to a lesser extent. The optimal conditions for the mentioned process were found to be: a degree of fineness of particles from 0.05 to 0.125 mm at [NaOH] = 19 mol/L, a solid-to-liquid ratio = 1:4 and a contact time of 3 h at 160 °C.

Composition	Content, wt%
Ln ₂ O ₃	67.478
P2O5 *	23.692
ThO ₂	6.921
Fe ₂ O ₃	0.522
Al_2O_3	0.077
SiO ₂	0.195
TiO ₂	0.144
CaO	0.631
MgO	0.135
U_3O_8	0.179
WO ₃	0.001
ZrO_2	0.014
ZnO	0.003
CdO	0.003
PbO ₂	0.002

Table 1. The average composition of monazite concentrate located at the UralMonatsit State Institution near the town of Krasnoufimsk, Sverdlovsk Region. * = calculated value.

At the end of the experiment, the autoclave was cooled to 100 °C, the obtained alkaline pulp was diluted with hot distilled water (1:1) and filtered. The precipitate was additionally washed from phosphate ions with water, dried, and further opened by the action of HNO_3 .

The chemical composition of the raw source material and precipitate after autoclave alkaline treatment was characterized using the X-ray phase ID analysis method using the DRON-4-13 diffractometer (Burevestnik Innovation Center of the Alrosa Group, RF, St. Petersburg, Russia). To identify the compound, the result was compared with diffractograms of the ICPDS catalog.

In the experiments on the acid dissolution of REE oxides/hydrated oxides, uranium elements and impurity elements (Zr, Si, etc.), samples weighing 1 g were placed in 100-mL quartz glasses, 5 to 50 mL of chemically pure 7–9 mol/L HNO₃ was introduced after heating to 80–95 °C for 1–5 h. In some experiments, 0.1 mL of 30% H_2O_2 was added to the solution. The insoluble part of the precipitate was dried, weighed, and the degree of opening of the sample (S, in wt%) was determined by the difference with the mass of the sample taken from the concentrate.

The activity of the resulting precipitates and solutions was measured using an SEG-1PPD semiconductor gamma spectrometer and a Quantulus 1220 alpha-radiation spectrometer. The total activity of the solutions was determined using a Progress-alpha alpha-radiometer. The relative error of gamma spectrometric was \pm 10%, alpha spectrometric, \pm 20%. The application of this technique made it possible to unambiguously identify and determine the content of ²³²Th, ²²⁸Th and ²²⁴Ra in the analyzed samples.

To isolate pure Th and U compounds from technological solutions, the authors applied the extraction method using tributyl phosphate solutions with different volume fractions of the extractant in the S-13 carbon diluent. The extraction part of the work was carried out in glass separation funnels. As extractants, solutions of chemically pure tributyl phosphate were used. Two model solutions were used in the experiment, containing 300 mg/L of U, 2500 g/L of Th, 10,000 mg/L of Ce, 5400 mg/L of La and 4500 mg/L of Nd (the total REE content was 19,900 mg/L) differing in the HNO₃ content as 3 and 1 mol/L, respectively. The experiments were carried out at room temperature, the ratio of the organic and aqueous phases upon contact was 0.1.

3. Results and Discussion

3.1. Studying the Distribution of Radioactive Impurities during Autoclave Leaching and Subsequent Processing

It is known that in the averaged composition of the Krasnoufimsk monazite sample, there is not only up to 67% of the sum of high-priced rare-earth oxides, but also 6.9% of thorium oxide ThO₂ and 0.18% of uranium oxides (Table 1). Thus, the initial mineral raw material is a radioactive material containing alpha emitters (232 Th, 235 U, 238 U) and their decay products (228 Th, 224 Ra, 226 Ra, 212 Bi, 220 Rn,

²²²Rn, and ^{212,214,216,218}Po), some of which are gamma-active (²²⁸Ac, ^{212,214}Bi, ²⁰⁸Tl, ²¹²Pb). The chain of radioactive transformations of ²³²Th is shown in Figure 1 diagram below:

$$^{232}Th \overset{228}{_{a}}Ra \overset{228}{_{b}}Ra \overset{228}{_{b}}Ac \overset{228}{_{b}}Th \overset{224}{_{a}}Ra \overset{220}{_{a}}Rn \overset{216}{_{a}}Po \overset{212}{_{b}}Pb \overset{208}{_{a}}Tl \overset{208}{_{a}}Pb \overset{208}{_{a$$

Figure 1. Chain of radioactive transformations of ²³²Th.

Therefore, before performing the technical operations of opening and extracting of REEs, uranium, and thorium from monazite concentrate, it is necessary to collect the necessary information on the distribution of radioactive impurities during processing and on the activity levels of liquid and solid phases.

In the course of the research, the following substances were subjected to radionuclide analysis: (1) precipitation of the obtained REE oxides/hydroxide compounds, elements of the uranium series and impurity elements; (2) a phosphate-alkaline solution; (3) a solution obtained by dissolving hydroxides in nitric acid; and also (4) the part of the solid phase that did not dissolve, which are formed in the technological process (Figure 2) [20].

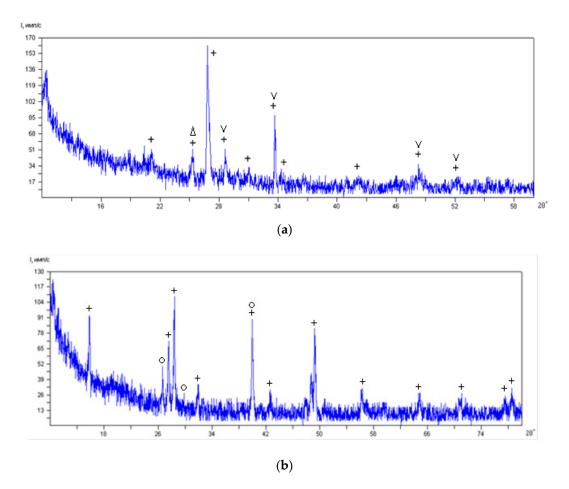


Figure 2. Diffractograms of (**a**) the initial sample of monazite concentrate; phase composition: "+" (Ce, La, Nd)PO₄, $Pr(PO_3)_3$; "v" UO₂; " Δ " TiO₂; (**b**) the product of alkaline autoclave opening of monazite concentrate under optimal process conditions; phase composition: "+" $Pr(OH)_3$, Ce(OH)₃, Nd(OH)₃; "o" Ce₂O₃.

It was found that, basically, all gamma and alpha-active radionuclides are concentrated in the deposits of REE oxides/hydroxides, elements of the uranium series, and impurity elements obtained as a result of alkaline treatment of monazite concentrate (Tables 2 and 3).

Table 2. Activity (kBq/L) determined by gamma spectrometry of samples: (a) oxide and hydrated precipitates of REEs, uranium and thorium obtained after autoclave processing of monazite concentrate; (b) phosphate-alkaline solutions and wash water; (c) nitric acid solution obtained by dissolving oxide and hydrated sediments of REEs, uranium and thorium.

Maternal	Radionuclide Used for	Activity, kBq/kg			
Radionuclide	Calculations	a	b	c	
	Ac-228	281	>0.1	5.2	
Th-232	Bi-212	285	Not found	Not found	
	Pb-212	265	>0.1	7.3	
	T1-208	229	>0.1	5.1	
U-238	Th-234	131			
	Pa-234	139	Below detection	Below detection	
	Bi-214	29	limit	limit	
	Pb-214	50			

Table 3. Activity (kBq/kg) determined by alpha spectrometry of the samples: (a) precipitation of oxide and hydrated rare-earth elements (REEs), uranium and thorium compounds obtained after autoclave processing of monazite concentrate; (b) phosphate-alkaline solutions and wash water; (c) nitric acid solution obtained by dissolving oxide and hydrated sediments of REEs, uranium and thorium.

Radionuclide	Th-232	Th-228	Ra-224	Rn-220	Po-212	Po-216	Σ
a	676	451	197	288	102	98	3470
b	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	-
с	4.5	5.3	2.3	3.5	1.6	1.7	31.6

The analysis of alpha and gamma spectra showed that the main identifiable radionuclides in the samples were ²²⁸Ac, ^{212,214}Bi, ^{212,214}Pb, ²⁰⁸Tl, ^{228,232,234}Th, ²¹²Po, ²²⁴Ra, and ²³⁴Pa.

On the contrary, in all experiments, the phosphate-alkaline solution obtained after the autoclave opening of the concentrate did not contain any radioactive impurities, as evidenced by the results of both gamma and alpha spectrometric measurements. In this regard, it can be concluded that the phosphate-alkaline (clarified) solutions formed in the processing, as well as the wash water, are free from radionuclides of uranium, thorium, and radium; they do not require special treatment and can be used to obtain a by-product, phosphate fertilizers.

It was shown that the solution obtained by dissolving hydroxides and hydrated oxides of REEs, uranium, thorium hydroxides in nitric acid contained impurities of radium-224 (Tables 2 and 3). However, due to its short half-life (about 4 days), the purification of the nitric acid phase from radium-224 is not expedient. In the authors' opinion, the sludge that was not dissolved in HNO₃ in terms of the content of radionuclides can be classified as industrial waste that does not require separate burial.

3.2. Opening of Monazite Concentrate

The heterogeneous high-temperature process of the hydroxide opening of monazite concentrate is carried out according to the general scheme:

$$(\text{REE})\text{PO}_4 + 3\text{NaOH} = (\text{REE})(\text{OH})_3 + \text{Na}_3\text{PO}_4 \tag{1}$$

$$Th_3(PO_4)_4 + 12NaOH = 3Th(OH)_4 + 4Na_3PO_4$$
(2)

In this regard, according to the above scheme and taking into account the optimal conditions recommended in the literature [28] for the complete conversion of monazite concentrate into the form

of hydroxides and hydrated metal oxides in an autoclaved alkaline environment, it was necessary to determine: (1) the optimal conditions for the quantitative washing of the resulting hydroxide precipitates from Na₃PO₄ and NaOH after autoclave processing; (2) the optimal conditions for the dissolution of hydroxide pulp in nitric acid; (3) approaches to the selective isolation and separation of target components: REEs, uranium, and thorium from the substance purified from natural radionuclides by the extraction method.

In the first stage, the conditions for washing hydroxide precipitates obtained after the autoclave opening of monazite concentrate from the remainders of NaOH and phosphate ions, primarily Na₃PO₄, were studied. During the tests, it was experimentally observed that the quantitative washing of 10 g of the product of opening the monazite concentrate from both NaOH and Na₃PO₄ to the content of phosphate ions in the wash water $< 10^{-3}$ g/L is achieved by a single wash with 100 mL of distilled water at 80 °C and contact time of 10 min. Trisodium phosphate in the environment of the alkali remaining after processing (up to 30%) at 100–110 °C easily crystallizes and is separated by a "blue ribbon" filter. In this case, Na₃PO₄ is isolated separately as a valuable by-product as a mineral fertilizer, and residual sodium hydroxide can be reused in experiments.

The second stage of research was focused on determining the optimal conditions for the dissolution of the obtained precipitation of hydroxides and hydrated oxides of REEs, uranium, and thorium in nitric acid after the autoclave processing of the concentrate.

To this purpose, the samples of monazite concentrate were autoclaved with alkali according to the procedures described in the previous section.

In the X-ray diffraction patterns of alkaline autoclave products there are no peaks assigned to the initial phases in the monazite concentrate, which indirectly confirms the completion of its opening (Figure 2).

The washed and dried samples of hydroxides and hydrated oxides of REEs, uranium, and thorium were dissolved in HNO₃ depending on acid concentration, temperature, Solid/Liquid phase ratios (S/L), and their contact times. The resulting solubility is shown in Table 4.

It appears that the main factors affecting the completeness of the dissolution of hydroxides and hydroxide precipitates of the concentrate components were the contact time and the concentration of HNO_3 . The S/L ratio and temperature values also affect the completeness of the dissolution of precipitation, but to a lesser extent. Under optimal conditions, the degree of opening of monazite concentrate reached values of up to 90–95%. On the contrary, a decrease in the contact time, S/L, and temperature, as well as an increase in the degree of fineness of the substance, reduces the degree of alkaline opening by 20–40%.

The introduction of an activating reagent H_2O_2 in the range from 0.01 to 0.1 mol/L does not significantly affect the degree of opening of the concentrate.

Based on the conducted research, optimal conditions for opening precipitates of rare-earth hydroxides, uranium, and thorium in HNO_3 are proposed: for 1 g of substance, 50 mL of 9 mol/L HNO_3 , and heating for 3 h at 90 °C without adding hydrogen peroxide.

In the third stage, two approaches were proposed for the extraction of uranium, thorium, and REEs from nitric acid solutions during the processing of monazite concentrate.

Quantitative extraction separation of REEs, uranium, and thorium with the simultaneous purification of impurity chemical elements is a non-trivial problem. The main known technologies for opening monazite raw materials were focused on the complete separation of the target rare-metal products, while the separation of fractions of elements of the uranium series and thorium was a necessary stage guaranteeing radioactive purification from REEs [30].

In the present study, two approaches were considered that were directly oriented to the extraction of thorium with a high yield of extraction from the mixture as a promising component of the thorium-uranium fuel cycle. The first approach was an attempt to move the components into two selective streams: a uranium stream and a stream of REEs and thorium. In the second approach, the separation of the components was carried out by co-separation of uranium and thorium into one selective stream followed by separation, and REEs into the second selective stream, followed by purification of rare-metal products by known methods at the second stage of the process.

No	T (°C)	<i>t</i> (h)	[HNO ₃] (M)	Solid/Liquid (g/mL)	S (%)
1			7.0	1:25	65.9
2			8.0	1:25	66.2
3		1.0	9.0	1:10	66.9
4			9.0	1:25	69.0
5			9.0	1:50	69.1
6	80		7.0	1:25	69.5
7		2.0	8.0	1:25	69.8
8			9.0	1:25	72.1
9			7.0	1:25	71.5
10		3.0	8.0	1:25	74.5
11			9.0	1:25	76.4
12			7.0	1:25	76.2
13		1.0	8.0	1:25	79.2
14			9.0	1:25	82.0
15			7.0	1:25	79.2
16		2.0	8.0	1:25	83.2
17	90		9.0	1:25	85.0
18			7.0	1:25	85.2
19			8.0	1:25	94.1
20		3.0	9.0	1:10	95.5
21			9.0	1:25	92.1
22			9.0	1:50	90.2
23			7.0	1:25	82.7
24			8.0	1:25	83.1
25		1.0	9.0	1:10	83.3
26			9.0	1:25	86.2
27			9.0	1:50	88.8
28	95		7.0	1:25	83.4
29		2.0	8.0	1:25	84.9
30			9.0	1:25	85.7
31			7.0	1:25	85.0
32		3.0	8.0	1:25	90.1
33			9.0	1:25	94.9

Table 4. The dependence of the solubility of precipitation of hydroxides of REEs, Th, U on the dissolution conditions. Sample weight 1 g.

The first approach included the extraction of uranium into a separate product, followed by purification from thorium-containing impurities by re-extraction with a complexing reagent. In the stage of uranium separation, 5% tributyl phosphate solutions were used as the extraction mixture, and 50% tributyl phosphate in S-13 at the stage of REE and thorium extraction. Data on the distribution of the components of the model solution are given in Table 5.

Table 5. The content of REEs, uranium and thorium at various stages of the technological scheme (in mg/L) in the first approach.

	Initial Solution	Raffinate 1	Reextrac U	Raffinate 2	Reextrac Th
La	5400	5400	2	5390	157
Ce	10,000	10,000	4	10,000	450
Nd	4500	4500	1	4490	360
Th	2500	2120	214	< 0.1	2100
U	30	1.31	29	< 0.1	< 0.1

It appears that when the proposed extraction scheme for separation and purification of REEs, uranium, and thorium is applied, and when the "5% tributyl phosphate—S-13" system is used, up to

90% U and up to 9.0% Th are extracted into the uranium branch. At the second stage of the process, the joint extraction of thorium and REEs into 50% tributyl phosphate in S-13, up to 80% with REE impurities in the organic phase was extracted into the selective thorium stream. In the aqueous phase, after the extraction of thorium, up to 100% of REEs remain in the raffinate. Thus, using the proposed scheme, it is possible to extract up to 90% of thorium from both streams.

The second approach is based on the joint quantitative separation of the sum of uranium and thorium into a separate branch, followed by a separation of the elements by re-extraction with diluted nitric acid under an elevated temperature (30–40 °C), which is traditionally used for the re-extraction of uranium in the Purex process [31]. It was shown that the joint extraction of uranium and thorium into 30% tributyl phosphate in S-13 with their subsequent joint isolation for re-extraction operations allowed the extraction of 90.74% of thorium and 78.49% of uranium (Table 6). The REE losses are not more than 4%.

	Initial Solution	Raffinate	Reextract U + Th
La	5400	5400	199
Ce	10,000	10,000	390
Nd	4500	4500	369
Th	2500	390	2270
U	30	< 0.1	23

Table 6. The content of REEs, uranium and thorium at various stages of the technological scheme (in mg/L) in the second approach.

4. Conclusions

In the development of the thorium-uranium fuel cycle, the optimal conditions of the technology for the extraction of thorium from monazite concentrate were identified, including the processes of grinding, opening of mineral raw materials, transfer of target components into a solution, followed by extraction, separation, and purification of uranium, thorium, and REEs.

In particular, the following optimal conditions for opening the monazite concentrate were determined: the action of 19 mol/L NaOH on mineral raw materials ground to particles ranging in size from 0.05 to 0.125 mm at S/L = 1:4 and the contact time of 3 h at 160 °C. The maximum process yield reaches 96 \pm 2%. The resulting phosphate-alkaline solution is free of radionuclides and can be used to produce Na₃PO₄ as a mineral fertilizer.

It was shown that the quantitative washing of the products of the alkaline opening of monazite concentrate from the residues of NaOH and Na₃PO₄ to the content of phosphate ions in wash water $<10^{-3}$ g/L is achieved with a single 10 min contact with distilled water at a temperature of 80 °C and S/L = 1:10.

Two versions of the extraction of thorium from its nitric acid solutions containing REEs and uranium were developed. The first variant starts with the separation of thorium together with REEs from uranium impurities due to its transfer to a 5% solution of tributyl phosphate in S-13. Further extraction of thorium from REEs is carried out by the extraction method using a 50% solution of tributyl phosphate in S-13. In the first variant, it becomes possible to extract up to 90% of thorium, up to 90% of uranium and up to 100% of REEs. The second option proposes the joint extraction of uranium and thorium into a 30% tributyl phosphate solution in S-13, followed by elemental separation during the re-extraction operations with the extraction of up to 90% of thorium and up to 78% of uranium. Up to 96% of REEs remain in the aqueous phase during extraction.

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