## PURIFICATION OF LANTHANUM OXIDE BY MAGNETIC NANO-COMPOSITE ALGINATE BEADS AFTER ITS CONCENTRATION FROM ACIDIC MONAZITE LEACH LIQUOR

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## ABSTRACT

Lanthanum represents nearly fifth of Egyptian mineral monazite total rare earth content. Due to its great industrial importance in applications, the present research is directed to its concentration to the highest possible grade then, its purification from the other minor coherent neighbor rare earth elements (REEs). Rare earth hydrated oxide cake was produced from Egyptian crude monazite mineral (54% purity, -125 mesh size) after its digestion with concentrated sulfuric acid, its dissolution with water, thorium separation, purification from uranium and then group precipitation of total rare earth elements as ammonium sulphates. Total REEs precipitate was transformed to its hydrated oxide form after its caustafication process. REEs hydrated oxide cake is subjected to process of cerium oxidation and separation. REEs cake after ceria separation contains nearly half of its content from lanthanum and subjected to air-ammonia precipitation scheme at different pH values to get a higher lanthanum concentrate, which reach to 78.6% purity with 85.5% recovery at pH5.8. Magnetic alginate beads are used as biosorbent for sorption of lanthanum (III) from an aqueous medium of pH 5.8. The magnetic alginate sorbent is one of the highest adsorption capacities, which equal 1.8 mmol  $g^{-1}$  or 0.25g/g. Column of 57 cm length, 1.2 cm diameter with flow rate 1ml/min experiment was carried out for adsorption of lanthanum which reaches to more than 99.68% purity with 80.7% recovery after elution with 1N HCl and 1N CaCl<sub>2</sub> at pH 3.5-4.

Keywords: Magnetic alginate; Monazite mineral.

#### **1. INTRODUCTION**

Monazite is a phosphate mineral with a rare earth content of approximately 70% rare earth oxide, primarily Ce, La, Pr and Nd (varies by deposit between 83.55% and 94.5% of total REO). It would consequently be considered a selective, cerium type mineral (Cesbron, 1986; Ferron et al., 1991; Gupta and Krishnamurthy, 1992, 2005; Trifonov, 1963). Monazite also includes 4–12 wt. % thorium and a variable amount of uranium (values are typically small but have been reported to be as high as 14 wt. %) (Gramaccioli and Segalstad, 1978; Gupta and Krishnamurthy, 2005).

Recently, lanthanides have the great demand for various functional materials (Kondo and Kamio, 2002). From the rare earth elements (REEs), lanthanum has special attention due to its technological importance because of increasing demands for advanced new materials. Current applications of lanthanum as a pure element or in association with other compounds are in super alloys, catalysts, and special ceramics and in organic synthesis (Palmieri et al., 2002). Many conventional methods for removing heavy metals, radionuclides and lanthanides from aqueous solution exist (Bhattacharyya and Sharma, 2004). These include adsorption, ion exchange, solvent extraction, coagulation, floatation, hyper-filtration, chemical precipitation, and reverse osmosis. All these conventional methods have some advantages as well as disadvantages, such as disposal of chemicals, resins and adsorbents to the nature may themselves be harmful.

Cost effective and sustainable production of high purity rare earth elements (REE) by solvent extraction and ion exchange techniques, but there are some problems of solvent extraction very complex, needs many steps to separate the REE ions in a high purity and causes a lot of environmental problems. In ion exchange process using resins as activated carbon (phosphorus modification and oxygen modification), Amberlite® and Lewatit® for the extraction of lanthanides as Nd, Ce, Er, La and Dy (Anja Grohme et al, 2014).

Synthesized resins have also been used in extraction of REEs. (M. Draye et al, 2000) described the alkaline polycondensation of 8-hydroxyquinoline with formaldehyde to obtain a polymeric resins. The selective uptake of  $Eu^{3+}$  from aqueous solutions containing  $La^{3+}$  was found to be due to the incorporation of 8-hydroxyquinoline in the molecular matrix of the phenolic resins.

Alginate, a natural high molecular-weight biopolymer composed of chains of 1, 4linked  $\beta$ -D-mannuronic and  $\alpha$ -L-guluronic has attracted much attention because it is non-toxic, selective, efficient and inexpensive (Da Costa and Leite, 1991; Chen and Wang, 2001; Chong et al., 2000). It has been demonstrated that the alginates are capable of binding metal ions through carboxyl groups (Siegel and Siegel, 1973; Lim et al., 2008) and calcium alginate beads are usually prepared for their gel forming properties in the presence of multivalent cations (Martinsen et al., 1989).

Magnetic separation technique, using magnetic polymeric particles, is a quick and easy method for sensitive and reliable capture of inorganic or organic pollutants. The magnetic sorbents behave similar to or even better than various commercial adsorbents (Ngah et al., 2006). After the usage, the magnetic sorbent can be easily separated from the solution by simple magnetic force. In the literature, iron oxides have been found to be successfully used as composite materials with host materials in fabricating magnetic sorbent (Oliveira et al., 2003, 2004; Chang and Chen, 2006).

Lanthanum sorption using magnetic alginate was studied at different experimental conditions. The sorption is highly dependent on the solution pH. Magnetic alginate sorbent is characterized by efficient sorption toward La(III) ions from aqueous medium at approximately pH 4. The uptakes kinetic were described by the PSORE, while the distribution of the metal at equilibrium between the solid and the liquid is modeled by the Langmuir equation. The maximum capacity reached 1.8 mmol g<sup>-1</sup> at 25°C. The sorption process was found to be endothermic, spontaneous (especially at high relative temperature) and contributes to increase the randomness of the system. Magnetic alginate beads regeneration was performed using 0.1 M CaCl<sub>2</sub> without loss of durability over six cycles. The regeneration efficiency was found to be in the range 89–93% (Khalid Z. Elwakeela, et al, 2017).

In this paper, the extraction of lanthanum is performed by a functionalized magnetic solid. Solid phase extraction (SPE) is an attractive method that offers advantages of

simplicity, flexibility, absence of emulsion, high enrichment factor, and is environmental friendly. The nature and the properties of the sorbent materials play an important role in the effective retention of analytes from the sample solution (Baghban N., et al, 2012).

## 2. EXPERIMENTAL

## 2.1. Chemicals

Sodium alginate with a molecular weight of 70,000–80,000 was purchased from Himedia (India). Fe(III) and Fe(II) salts, citric acid, calcium chloride, Hydrochloric acid, sulfuric acid, for solution preparations, double-distilled water was used. The pH of the solutions was adjusted to the required value with 0.01 mol/L HCl and 0.01 mol/L NaOH solutions. All chemicals used with analytical grade.

## 2.2. Analysis

The most common analytical techniques used to determine lanthanum in monazite leach liquor and in the products produced by Ion Chromatograph (IC-UV-VIS) instrument (DX-500) manufactured by Dionex Corp., USA, ICP-OES (Prism ICP- high dispersion) (Teledyne leeman labs) and X-ray fluorescence (EDXRF) using Rigaku spectrometer NEX CG with polarized optics in EDXRF lab, pH meter (inoLab, pH Level 1, WTW, Germany), Nuclear Materials Authority, Cairo.

## 2.3. Procedure

## 2.3.1. Synthesis of magnetic alginate

Stoichiometric amount of Fe(III) and Fe(II) salts was dissolved in distilled water under vigorous stirring. Then, a solution of NaOH was added very slowly until the pH value was raised to  $\sim 11$  at which a black suspension was formed. This suspension was then refluxed at 80°C for 4.0 hours under vigorous stirring and N<sub>2</sub> gas.

Magnetic particles were separated from the aqueous solution by centrifugation, washed with double distilled water several times, and then dried in an oven over night. Coated particles were prepared by adding nano-magnetite to citric acid solution (2.0% w/v) and then the mixture was stirred for 30 minutes at 80°C.

After precipitation, the coated particles were dispersed in water to obtain a stable ferrofluid. 50 mL of ferrofluid was added very slowly to mixing of 12 g of sodium alginate powder with double distilled water (4.0% w/v solution) at 50°C. The mixture was further stirred for 3 hours to ensure the homogeneity of the system.

The obtained viscous solution was introduced dropwise using a syringe into 400 mL of  $CaCl_2$  bath (5.0% w/v). Regeneration of the magnetic alginate sorbent was carried out using 0.1 M CaCl\_2. Magnetic alginate beads have a good performance for repeated use up to at least six cycles. (Khalid Z. Elwakeela, et al, 2017). The structure of alginate was shown in figure 1 and figure 2 shows the structure after the reaction with CaCl\_2. (H. Abd El Monem, 2017).



Fig. 1: Molecular structure of alginate macromolecules and the reaction of the formation of alginate nano-composite beads.



Fig. 2: Molecular structure of the prepared magnetic alginate.

#### 2.3.2. Preparation of the acid leach liquor of monazite

Monazite from black sands was used as raw material for the experiments. Monazite of 54% grade with -125 mesh size was digested with concentrated sulfuric acid with (1:2) ratio. Hence, 100 g of monazite mixed with concentrate sulfuric acid, stirring for 2.5 hrs at 200-220°C, and then diluted with iced water. The slurry was filtrated and the filtrate was diluted with water to 2L (original solution), and the remaining residue constitutes silica, zircon and other undigested gangue minerals (Abd Elfatah, 2012).

Three main reactions occurring in the mineral digestion reaction as follows:

$2 \operatorname{REPO}_4 + 3 \operatorname{H}_2 \operatorname{SO}_4 \longrightarrow$	$(RE)_2(SO4)_3 + 2 H_3PO_4 \dots (1)_3 H_3PO_4 \dots $	)
$Th_3(PO_4)_4 + 6 H_2SO_4 \longrightarrow$	$3 \text{ Th}(\text{SO}_4)_2 + 4 \text{ H}_3 \text{PO}_4 \dots (2$	2)
$U_3(PO_4)_4 + 6H_2SO_4 \longrightarrow$	$3 U(SO_4)_2 + 4 H_3PO_4$	3)

#### 2.3.3. Thorium separation

One litre from the original solution was taken to precipitate thorium at pH 1.3 by ammonia solution (1:1), where thorium was precipitate as thorium pyrophosphate as the following equation:

$$Th(SO_4)_2 + 2 H_3PO_4 \longrightarrow ThP_2O_7 + 2 H_2SO_4 + H_2O \dots (4)$$

After thorium separation, the pH of the filtrate was adjusted by sodium hydroxide to 1.75 and filter again. The produced filtrate was then go to uranium decontamination step.

#### 2.3.4. Uranium decontamination

The produced filtrate was passed through a column of Amberlite IR 402-Cl with flow rate 1ml /min and bed height 16 cm. The resin absorbs uranium from filtrate and passes the rare earth elements through the solution, which could be described by the following reaction (K. A. Rabie, 2016).

$$[UO_{2} (SO_{4})_{3}(H_{2}O)_{4}]^{-4} + (4+n) HAmb \longrightarrow [UO_{2}(SO_{4})_{3}(H_{2}O)_{2}]^{-4}(HAmb)n + 2[H_{3}O]^{+} \dots (5)$$

#### 2.3.5. Hydrated oxide (hydroxide) preparation

The produced solution from the column of resin (effluent) was precipitated by sodium hydroxide over stoichiometric ratio at pH 9 then wash several times with water till neutralization.

$$RE_2(SO_4)_3 + 6 NaOH \longrightarrow 2 RE(OH)_3 + 3 Na_2SO_4 \dots (6)$$

#### 2.3.6. Cerium separation

Drying the produced hydrated oxides of rare earth elements at 110°C then oxidized at 160°C for 10 hours with airflow to oxidize cerium (III) to less basic cerium (IV). The cake was leached with HCl (1N) for 1 day, where the leaching is to dissolve the trivalent REEs leaving the less basic ceric oxide undissolved.

## 2.3.7. Lanthanum concentration

After the removal of the unreacted part (solid ceric part), take the filtrate and a differential precipitation was performed first at pH 3 and finally at pH7 by ammonia solution in the presence of air. The final filtrate was precipitated at pH 9 by ammonium carbonate, which produces a concentrate of lanthanum precipitate.

## 2.3.8. Reaction of lanthanum with magnetic alginate resin

After precipitation at pH 5.8 and before the final precipitation by carbonate, 700 ml of the filtrate was withdrawn and passes through a column of the magnetic alginate beads with length of 57 cm and 1.2 cm diameter with flow rate of 1 ml/min.

Cross-linked metal-alginate complexes are formed when the blocks of alginate macromolecular chains are mediated by trivalent metal ions. The exchange process takes place by the replacement of  $Ca^{+2}$  counter ions in the alginate macromolecule by trivalent metal ions from the external electrolyte solution. This exchange occurs inherently and stoichio-metrically as shown in equation 7.

## **3. RESULTS AND DISCUSSION**

## **3.1.** Concentration of Lanthanum

## 3.1.1. Analysis of acidic monazite leach liquor

The sulphuric acid process involves the digestion of monazite sands using hot concentrated sulphuric acid to form a monazite sulfate paste of rare earths and the remaining residue consists mainly zircon. The digestion reaction results a thick gray paste, so after cooling, it can be treated with about 20 parts of cold water to one part by weight of digested sand and leached for two hour, then filtration process. The produced acidic monazite leach liquor contains mainly 2.8% ThO<sub>2</sub>, 0.22% U<sub>3</sub>O<sub>8</sub> and 29% total RE<sub>2</sub>O<sub>3</sub>. The individual ppm of each element is illustrated in table 1.

Rare earth elements	<b>Concentration (ppm)</b>
$ThO_2$	1400
$U_3O_8$	110
La <sub>2</sub> O <sub>3</sub>	3480
Ce <sub>2</sub> O <sub>3</sub>	6960
$Pr_2O_3$	870
$Nd_2O_3$	2030
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	1015

 Table (1): Chemical analysis of clear sulphates leach liquor of monazite.

#### 3.1.2. Thorium separation

From the experimental work, after monazite digestion, dissolution and filtration, the acidic leach liquor was firstly precipitated at pH 1.3 to separate thorium as thorium pyrophosphate. The concentration of the acidic solution after complete thorium separation process was summarized in table 2.

Rare earth elements	<b>Concentration (ppm)</b>	% Recovery*
$ThO_2$	Un-detected	
$U_3O_8$	107.5	97.73
$La_2O_3$	3470	99.70
$Ce_2O_3$	6530	93.82
$Pr_2O_3$	850	97.70
Nd <sub>2</sub> O <sub>3</sub>	2025	99.75
Middle and heavy $RE_2O_3$	990	97.54

Table (2): The chemical analysis of monazite acidic
solution after thorium separation.

\* % recovery calculated related to the first acidic liquor elemental concentration in table 1.

#### 3.1.3. Uranium decontamination

After thorium separation, the filtrate was secondly precipitated at pH 1.75, which considers the suitable pH for passing through the resin; uranium was adsorbed at this pH. The produced filtrate was passed on Amberlite IR402Cl column, which adsorb uranium ion, the produced raffinate contains most of rare earth concentrate.

Rare earth elements	<b>Concentration (ppm)</b>	% Recovery*
$ThO_2$	Un-detected	
$U_3O_8$	Un-detected	
$La_2O_3$	3465	99.57
$Ce_2O_3$	6505	93.46
$Pr_2O_3$	840	96.55
Nd <sub>2</sub> O <sub>3</sub>	2020	99.51
Middle and heavy $RE_2O_3$	980	96.55

Table (3): The chemical analysis of acidic solution after uranium separation.

\* % Recovery calculated related to the first acidic liquor elemental concentration in table 1.

#### 3.1.4. Cerium separation

The solution that passes from the column was caustaficated by sodium hydroxide, and then the produced precipitate was washed several times till neutral. The hydrated rare earth oxide was dried and oxidized in air with convection at 160°C for 10 hours to oxidize cerium III to less basic cerium IV, which was Leached by dilute hydrochloric acid (1N) over night. The unreacted part contain 90% of less basic Ce<sub>2</sub>O<sub>3</sub> and 10% Ln<sub>2</sub>O<sub>3</sub>, the analysis of cerium concentrate is demonstrated in table 4. The product of cerium separation stage was declared in figure 3.

Rare earth elements	<b>Concentration (ppm)</b>	% Recovery*	% Purity
La <sub>2</sub> O <sub>3</sub>	346.5		
Ce <sub>2</sub> O <sub>3</sub>	5854.5	84.12	88.91
$Pr_2O_3$	84		
Nd <sub>2</sub> O <sub>3</sub>	202		
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	98		

Table (4): The chemical analysis of cerium concentrate.

\* % Recovery calculated related to the first acidic liquor elemental concentration in table 1.



## Fig. 3: Cerium separation process by leaching with 1N hydrochloric acid.

The analysis of the produced acidic leach liquor after filtration to remove the unreacted less basic ceric oxide was gathered in table 5.

Rare earth elements	<b>Concentration (ppm)</b>	% Recovery*	% purity
$La_2O_3$	3118.5	89.61	43.16
$Ce_2O_3$	650.5		
$Pr_2O_3$	756	86.90	
$Nd_2O_3$	1818	89.56	
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	882	86.90	

 Table (5): The chemical analysis of acidic solution after cerium separation.

\* % Recovery calculated related to the first acidic liquor elemental concentration in table 1.

## 3.1.5. Air-ammonia precipitation for heavier REEs than lanthanum

After cerium separation, the acidic leach liquor was differentially precipitated at different pH values from 3 to 7 by air with ammonia solution. The analysis of each precipitate was illustrated in table 6.

 Table (6): The chemical analysis of precipitates after applying air-ammonia precipitation scheme at different pH values.

			=		
рН	3	4	5	6	7
Rare earth elements		Conce	entration (	ppm)	
$La_2O_3$	31.19	62.37	124.74	249.48	1247.4
$Ce_2O_3$	325.25	455.35	520.40	585.45	617.98
$Pr_2O_3$	189	302.4	453.6	642.6	718.2
$Nd_2O_3$	545.4	727.2	1090.8	1545.3	1745.28
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	352.8	441	661.5	793.8	864.36

After precipitation at different pHs, the precipitated values of each element were increased by pHs increase. The precipitation at pH 5.8 was considered the most promising one because it precipitates most of the heavier rare earth elements than lanthanum with lesser lanthanum loss. The analysis of this precipitate is presented in table 7.

Rare earth elements	<b>Concentration (ppm)</b>	% Recovery*
$La_2O_3$	156	
$Ce_2O_3$	580	
$Pr_2O_3$	580	69.05
$Nd_2O_3$	1370	67.82
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	760	77.55

Table (7): The chemical analysis of precipitate at pH 5.8after applying air-ammonia scheme.

\* % Recovery calculated related to the first acidic liquor elemental concentration in table 1.

The analysis of the produced leach liquor after filtration by air-ammonia method at pH 5.8 for lanthanum concentration was illustrated in table 8. The product of air-ammonia precipitation process is shown in figure 4.

Table (8): The chemical analysis of the acidic liquor after applyingair-ammonia precipitation scheme at pH 5.8.

Rare earth oxides	<b>Concentration (ppm)</b>	% Recovery*	% purity
$La_2O_3$	2962.5	85.5	78.56
$Ce_2O_3$	70.5		
$Pr_2O_3$	176		
Nd <sub>2</sub> O <sub>3</sub>	440		
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	122		

\* % Recovery calculated related to the first acidic liquor elemental concentration in table 1.



Fig. 4: Lanthanum separation (concentration) by air-ammonia method.

After reaching these results, the steps were applied to the rare earth separation project, and the design of the applied unit for lanthanum separation was illustrated in figure 5. The prospected and applicable air- ammonia precipitation method was recommended to produce commercial lanthanum concentrate with higher purity and recovery at selected pH values which can be improved in a future lab works to be scaled up for industrial application.



Fig. 5: Applied first design for the probable separation and concentration of commercial La.

## 3.2. Purification of Lanthanum by Using Magnetic Alginate Resin

After precipitation at pH 5.8, the resulted filtrate was passed through a column of magnetic alginate resin of 57cm length, 1.2cm diameter with flow rate of 1ml/1min (complete loading time was nearly 12 hours for 750 ml loaded filtrate).

The analysis of the raffinate produced (totally collected) from column as shown in table 9. The elution was carried out by 750 ml of  $1N \text{ HCl} + 1NCaCl_2$  at pH 3.5-4 in 12 stages (each cut was taken after 60 minutes) with the same rate of loading.

**Rare earth elements Concentration (ppm)** % Recovery Inside the column  $La_2O_3$ Un-detected 85.5  $Ce_2O_3$ Un-detected \_\_\_\_\_ Un-detected  $Pr_2O_3$ \_\_\_\_\_  $Nd_2O_3$ Un-detected \_\_\_\_\_ Middle and heavy  $RE_2O_3$ 32 \_\_\_\_\_

Table (9): The chemical analysis of the raffinate produced from the resin column.

Eluting Stages	1	2	3	4	5
Rare earth elements	Concentration (ppm)				
$La_2O_3$	31	28	44	63	138
Ce <sub>2</sub> O <sub>3</sub>	25	15	16	10	4
Pr <sub>2</sub> O <sub>3</sub>	64	43	41	25	5
Nd <sub>2</sub> O <sub>3</sub>	85	164	179	8	Un-detected
Middle and heavy $RE_2O_3$	59	24	5	Un-detected	Un-detected

Table (10): The analysis of the eluting solution of the 1<sup>st</sup> five successive stages.

After the elution of the fourth stage, we adjust the pH of the eluting solution to less than 3 by HCl to ensure the release of the rest of the element, after ensuring that the solution is free of most other elements.

All the eluting products were collected from step 5 through 12 for the production of pure lanthanum. The analysis of the collected eluting solutions was represented in table 11.

Table (11): The chemical analysis of the collected eluting solution fromthe 5 to 12 stage for pure lanthanum production.

Rare earth elements	Concentration (ppm)	% Recovery*	% Purity
$La_2O_3$	2806.5	80.65	99.68
Ce <sub>2</sub> O <sub>3</sub>	4		
$Pr_2O_3$	5		
Nd <sub>2</sub> O <sub>3</sub>	Un-detected		
Middle and heavy RE <sub>2</sub> O <sub>3</sub>	Un-detected		



Fig. 6: Applied mass balanced flow diagram for separation & purification of lanthanum.

## **4.** CONCLUSION

The Egyptian beach monazite mineral was digested in its impure native form (~54% purity and -125 mesh size) by concentrated sulphuric acid. The produced paste was dissolute in cold water to sulphates acidic leach liquor of the mineral cations.

Thorium was first separated completely by precipitation at pH 1.3 then rare earth elements consequently were precipitated at pH 1.75 by dilute ammonia then transformed to sulphate form and treated by resin to be decontaminated from uranium. The produced raffinate was precipitated by soda and oxidized to remove cerium as Ce (IV) leaving trivalent rare earth elements. The trivalent clear solution was subjected to air-ammonia precipitation scheme for lanthanum concentration. By adjusting at pH 5.8, most of trivalent rare earth elements were precipitated leaving lanthanum in the solution with 78.56 % purity and the percent of recovery reaches 85.5. Lanthanum concentrate solution was passed to a column of magnetic alginate beads of 57cm length, 1.2cm diameter with flow rate of 1ml/1min. The elution was carried out by 750 ml of 1N HCl + CaCl2 at pH 3.5-4 in 12 stages. Collect the eluting solutions from 5 to 12 stages for the production of lanthanum with more than 99.9% and 80.65 % recovery.

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# تنقية أكسيد اللانثانوم بواسطة خرز متراكبات النانو الجينات المغناطيسية بعد تركيزه من محلول المونازيت الحامضي

## **عزيز ة إبر اهيم لطفي عبد الفتاح** هيئة المواد النووية- القاهر ة- مصر

يمثل اللانثانوم ما يقرب من خمس المحتوى الكلى للعناصر الأرضية النادرة الناتجة من معدن المونازيت المصري. نظرا لأهمية العنصر الكبيرة على المستوى التطبيقي والصناعي، يوجه هذا البحث إلى تركيزه إلى أعلى درجة ممكنة ثم تنقيته من أقرانه من العناصر الأرضية النادرة. تم إنتاج عجينة الأكاسيد الرطبة من العناصر الأرضية النادرة الناتجة من معدن موناز أيت الخام المصري (54٪ نقاء، -125 ميش) بعد هضم المعدن مع حامض الكبريتيك المركز، وحله بالماء، وفصل الثوريوم، وتنقية اليور انيوم، ثم ترسيب للعناصر الأرضية النادرة مجتمعة والتي يتم تحويلها إلى أكاسيد رطبة عن طريق التفاعل مع الصودا الكاوية. يلى ذلك أكسدة وفصل السيريوم من محتوى أكاسيد العناصر مجتمعة. بعد فصل السيريوم نجد أن محتوى أكاسيد العناصر مجتمعة يحتوي على ما يقرب من نصف محتواه من اللانثانوم ويتم الترسيب بواسطة الأمونيا في وجود الهواء عند قيم مختلفة من الأس الهيدر وجيني للحصول على أعلى تركيز اللانثانوم التي تصل إلى نسبة نقاوة 78.6٪ ونسبة استرجاع تصل إلى 85.5% عند pH. يتم استخدام خرز الجينات المغناطيسية لامتصاص اللانثانوم ثلاثي التكافؤ من الوسط المائي عند pH. حيث أن حبيبات الجينات المغناطيسية لديها أعلى قدرة للامتزاز (1.8 مللي مول/جم راتنج أو 0.25 جم /جم راتنج). تم عمل عامود من الراتنج طوله 57 سم، قطره 1.2 سم مع معدل تدفق 1 مللي/1 دقيقة والذي تم من خلاله ادمصاص اللانثانوم والذي تم استرجاعه بنسبة نقاوة تصل إلى أكثر من 99.68% ونسبة استرجاع 80.7٪ باستخدام HCl + .4-3.5 عند در جة حموضة 1NCaCl<sub>2</sub>