



Article Hydrochloric Acid Leaching of Philippine Coal Fly Ash: Investigation and Optimisation of Leaching Parameters by Response Surface Methodology (RSM)

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Abstract: Coal Fly Ash (CFA) is a hazardous waste from coal-fired power plants, but has increasingly become a popular supplementary cementitious material for cement in the construction industry. As a secondary resource of REE, its main advantage lies in its fine particle size that eliminates the need for costly and energy-intensive comminution. In this study, the potential of CFA from the Philippines as a secondary REE resource was investigated by direct leaching of REEs with hydrochloric acid (HCl). The CFA sample came from a coal-fired power plant with a circulating fluidized bed combustion (CFBC) technology. For the leaching tests, the effects of HCl concentration, leaching time and leaching temperature on REE extractions were elucidate optimized via Response Surface Methodology (RSM). The RSM results showed that the optimum leaching occurred at 3M HCl, 65 °C and 270 min with extractions of Nd, Er, Eu, Tb and Dy at 70.8%, 76.34%, 88.02%, 90.01% and 73.38%, respectively. According to these results, the CFA from the Philippines is a promising secondary resource of REE and the extraction method employed was effective in achieving a relatively high REE dissolution. Moreover, the empirical model that was established accurately predicted the dissolution of REE with an accuracy of 98.20%, 96.66%, 97.09%, 98.17% and 97.78% for Nd, Er, Eu, Tb and Dy, respectively.

Keywords: REE; coal fly ash; RSM; beneficiation; recovery; leaching

1. Introduction

Rare earth elements (REE) are the 17 elements in the periodic table which comprise the 15 lanthanides plus scandium and yttrium. REEs are essential for many high-technology applications such as consumer electronics, military hardware, renewable energy and clean storage technologies. A three-megawatt wind turbine, for example, requires three tons of REEs [1]. With the global push to decarbonize society to combat climate change, the demand for REEs is projected to increase by 5–9% for the next 25 years [1]. However, the demand for REEs is outstripping its supply by about 3000 tons per year [2,3]. This supply-demand imbalance in the market is driving up the REE prices, which in turn is fueling the need to explore new REE resources and cost-effective extraction technologies [4–6].

Recent research findings found that coal fly ash (CFA) can be a potential alternative source of many elements including the valuable heavy REEs and other strategic elements [7–17]. CFA is a combustion by-product from burning of coal in coal-fired power plants characterized by its fine particle size and high amorphous content. On one hand, high amorphous content of CFA has made it popular in geopolymers and as supplementary



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Copyright: © 2022 by the authors. Licensee MDPI, Basel, Switzerland. This article is an open access article distributed under the terms and conditions of the Creative Commons Attribution (CC BY) license (https:// creativecommons.org/licenses/by/ 4.0/). cementitious material [2]. On the other hand, this waste material is hazardous and contains environmentally regulated contaminants like boron, selenium and arsenic [3,4]. Millions of tons of CFA is produced annually in countries such as South Africa, the USA, China, Australia, Greece, Japan and Poland [16]. In the Philippines, more than 1.4 M metric tons of CFA is generated annually [18] and this quantity is expected to increase by around 10% annually [19]. As of August 2021, there are 50 coal-fired power plants operating in the Philippines and several more are approved for construction in the coming years [20]. To reduce the volume of CFA disposed of in landfills and minimize environmental and disposal issues associated with this industrial waste, its utilization and valorization, especially as a potential source of rare earth elements, has been the focus of much research in recent years. Findings of studies conducted on CFAs from the USA, Poland and Russia showed that the average REE content in CFA is around 400-600 ppm, which is approximately three times higher than that of the conventional REE ore [8,9].

The conventional way of extracting REEs from many of its natural ores, including waste materials, is leaching. This method involves dissolution of the target components in a sample by lixiviants such as acids or bases. In the past, acid leaching has been used to extract hazardous elements from CFAs to decontaminate them before disposal [21–24]. More recently, these methods were improved to facilitate the selective extraction of valuable elements from CFA, such as Fe, Al and REEs [25–32]. Hydrochloric acid (HCl) is the lixiviant commonly used to extract REE from CFAs and is preferred over cheaper sulfurc acid because it does not form gypsum, a mineral that causes significant REE loss [30–33]. HCl is also cheaper than nitric acid (HNO3) and is better than sodium hydroxide (NaOH) because it limits silica dissolution that interferes with CFA dissolution and filtration through the formation of silica gel [15]. The investigation of the effects of HCl concentration, temperature, stirring speed, solid-to-liquid ratio and leaching time on the extraction of lanthanum (La), cerium (Ce) and neodymium (Nd) has already been conducted on Chinese CFA [31]. To further improve the extraction of the REEs, pretreatment methods are employed, such as magnetic separation, caustic cracking and roasting [33,34]. To recover the REEs from the leachates, adsorption is the primary existing method that is being used [35–37].

In this study, the combined effects of HCl concentration, temperature and leaching time were elucidated on the REE extraction from Philippine CFA using batch-type experiments. The set of parameters investigated in this study is quite similar to the previous study of Cao et al. [31] but particularly distinct in the following: (a) leaching time beyond 2 h was considered since according to King et al. [32] the HCl continues to leach from CFA up to 35 days; (b) the recovery of all six critical REEs was investigated, rather than just La, Ce and Nd only; (c) optimization of parameters using central composite design (CCD) was conducted in combination with Response Surface Methodology (RSM) leading to the development of a model based on the experimental design to determine the optimum leaching conditions for REE extraction, and; (d) Philippine CFA was used as the raw material. Different types of CFA have different responses to HCl [32] and the degree of surface deposition and accompaniment of REE in CFA is greatly affected by the type of the original coal and combustion conditions such as temperature, oxygen, concentration, residence time and cooling rate of CFA particles [29].

2. Materials and Methods

2.1. Material

The CFA used in this study came from a coal-fired power plant with a circulating fluidized bed technology. Grab sampling and Jones riffle was used on site to collect the samples, which were then sieved in the laboratory using a 120-mesh (125 microns) Tyler standard sieve. Samples that passed through the 120-mesh were dried in the oven for 48 h at 105 °C and then analyzed by scanning electron microscopy with energy dispersive X-ray spectroscopy (SEM-EDS, InTouchScopeTM, JSM-IT200, JEOL Ltd., Hokkaido, Japan) and X-ray Diffraction (PANalytical X'Pert MPD Pro, Hokkaido, Japan) to obtain the mineralogical characteristics. Elemental maps were also taken with SEM-EDS using an accelerating

voltage of 15 keV at 10,000 cps with a 1.0 ms time constant under ultrahigh vacuum. The chemical composition was determined by inductively-coupled plasma-mass spectroscopy (ICP-MS, Agilent 7700, Agilent Technologies, Manila, Philippines) (margin of error = $\pm 2\%$).

2.2. Methods

A series of experiments were conducted to determine the optimum dissolution of REEs from the CFA. For this purpose, leaching experiments were carried out using an improvised water bath with a 125-mL Erlenmeyer flask (Figure 1) containing 50 mL of HCl with 5 g of CFA. To make sure that the temperature was maintained during the entire leaching duration, the Erlenmeyer flask was immersed in water in an aluminum container placed on a hot plate and the temperature of the water was monitored using a thermometer. To prevent the escape of vapor, the Erlemeyer flask was fully covered and the solution was kept homogeneous by constant stirring with a magnetic stirrer. After subsequent leaching, the solutions were filtered through 0.45 μ m filters and the leachates were analyzed using ICP-MS (ICP-MS, Agilent 7700, Agilent Technologies, Santa Clara, CA, USA) (margin of error = $\pm 2\%$).



Figure 1. A schematic diagram of the leaching experiment set-up.

2.3. Multivariate Design of Experiment

It has been known that several independent variables can affect the dissolution of REEs in the CFA and it is therefore necessary to select those variables that make major contribution to the response variable through the screening of variables [38–43]. In this study, screening was carried out using a 3-factor factorial design to determine which among the independent variables (acid concentration, leaching time and leaching temperature) significantly contributed to the response variable and also to find out if there are interactions among the chosen parameters. The acid concentration, leaching temperature and leaching time were varied at four levels (0.25 M, 0.5 M, 1 M and 3 M), two levels (25 °C and 65 °C), and five levels (30 min, 60 min, 180min and 360 min), respectively. The stirring speed and the solid:liquid ratio were kept constant at 260 rpm and 1:10, respectively. The significance of the parameters and their interactions were analyzed using 3-way analysis of variance (ANOVA).

Based on the results of the screening test, the independent variables to be optimized as well as their corresponding levels were set-up using central composite design (CCD) response surface methodology (RSM) and the range of each factor of interest is presented in Table 1 [43].

Independent Variable	Symbol	Levels of Coded Variables						
		$-\alpha$	Low	Medium	High	+α		
		-1.6817	-1	0	1	+1.6817		
Acid conc (M)	X ₁	0.3	1	2	3	3.7		
Temp (°C)	X ₂	25	35	50	65	75		
Time (min)	X ₃	30	90	180	270	330		

Table 1. A central composite design RSM matrix for three design variables.

To minimize the error and effect of uncontrolled factors, the experimental runs were randomized. The order of the runs is summarized in Table 2 and the response was used to generate an empirical model based on a second-degree polynomial equation (Equation (1)):

$$Y = \beta_0 + \sum_{i=1}^n B_i X_i + \sum_{i=1}^n \beta_{ii} X_i^2 + \sum_{i=1}^n \sum_{j=1}^n \beta_{ij} X_i X_j + \varepsilon$$
(1)

where *Y* is the predicted response; β_0 is the constant coefficient; β_i are the linear coefficients; β_{ii} are the quadratic coefficients; β_{ij} are the interaction coefficients; *n* is the number of factors of interest; X_i and X_j are the coded values of the variable leaching parameters; and ε is the random error [42].

|--|

Standard Bun No	Dun	Independent Variable in Coded Form					
Standard Kun No.	Null	X ₁	X ₂	X ₃			
1	8	-1	-1	-1			
2	4	+1	-1	-1			
3	5	-1	+1	-1			
4	14	-1	-1	+1			
5	1	+1	+1	-1			
6	18	+1	-1	+1			
7	6	-1	+1	+1			
8	3	+1	+1	+1			
9	13	$-\alpha$	0	0			
10	12	+α	0	0			
11	2	0	$-\alpha$	0			
12	15	0	+α	0			
13	16	0	0	$-\alpha$			
14	7	0	0	$+\alpha$			
15	17	0	0	0			
16	10	0	0	0			
17	9	0	0	0			
18	11	0	0	0			

ANOVA was used to describe the interaction between the process variables and the response variable. The Design Expert version 11 Software by State-Ease Inc., Minneapolis, USA was used to obtain the following: (a) regression analysis of the experimental data; (b) statistical significance of independent variables through F-test; and (c) \mathbb{R}^2 coefficient to determine the accuracy of the fitted model. The significant model terms were evaluated by the probability value (*p*-value) at 95% confidence interval.

3. Results and Discussion

This section covers the discussion on the characterization of the CFA raw material and the extraction of critical REEs, particularly Dy, Nd, Er, Eu and Tb. There was no data obtained for Y because its presence was below the detection limit of the ICP-MS used in the analysis and hence it is not included in the discussion. The extraction of major elements Al, Fe, Ca and Fe are also presented in relation to the extraction of REEs.

3.1. Characterization of CFA

The REE composition of the CFA determined by ICP-MS analysis is summarized in Table 3. The elements are presented according to the economic classification of Seredin and Dai (2010) which is based on the current market trend of the individual REE rather than on the traditional classification of light REE as not valuable and heavy REE as valuable. This economic classification is the basis for the grading of the REE ore through an index called outlook coefficient which measures the ratio of the total critical REE to the total excessive REE, and the relative amount of the critical REE. An ore is regarded as promising REE raw material for economic development if its outlook coefficient is within 0.7 to 1.9 and the relative amount of critical REE is between 30% and 51% [15]. Based on this classification, the CFA used in this study qualifies as a promising REE raw material with an outlook coefficient of 1.03 and critical elements percentage of 33.3%.

Category	REE Element	Head Grade (ppm)
Strategic	Scandium (Sc)	16.67
	Lanthanum (La)	31.2
Uncritical	Gadolinium (Gd)	5.56
Uncritical	Praseodymium (Pr)	7.13
	Samarium (Sm)	5.43
	Dysprosium (Dy)	4.63
	Erbium (Er)	2.62
Critical	Europium (Eu)	1.36
Cifical	Neodymium (Nd)	26.8
	Terbium (Tb)	0.78
	Yttrium (Y)	28.0
	Cerium (Ce)	58.6
	Holmium (Ho)	0.90
Excessive	Lutetium (Lu)	0.35
	Thulium (Tm)	0.34
	Ytterbium (Yb)	2.19

Table 3. Relative abundance of REEs in the CFA.

Figure 2 shows the XRD pattern of the CFA samples and it can be seen that the dominant minerals present are quartz (SiO₂), calcite (CaCO₃), iron oxide minerals such as magnetite (Fe₂O₄), hematite (Fe₂O₃) and maghemite (γ -Fe₂O₃), and aluminosilicates such as illite ((K, H₃O)(Al, Mg, Fe)₂(Si, Al)₄O₁₀[(OH)₂, (H₂O)]) and biotite (K(Mg, Fe)₃(AlSi₃O₁₀)(F, OH)₂).

The quantity of the abundant minerals present was determined by XRF analysis. The results presented in Figure 3 show that SiO_2 is the most abundant mineral, followed by Al_2O_3 , Fe_2O_3 and CaO with small amounts of MgO, S, K_2O and TiO_2 . A high amount of silica may not be desirable for leaching due to the formation of a silica gel layer which prevents diffusion of ions. However, the presence of calcium ions can somewhat counteract the gelling effect of the silica through the reaction of calcium ions with silica forming

calcium silicates, thereby destabilizing the gel layer and reducing the amount of silicate [32]. The CaO present in the CFA has to be dissolved so that it can react with the silica, and based on thermodynamics, CaO is easily dissolved by HCl [44]. This is therefore one of the bases for the choice of HCl as lixiviant in this study.



Figure 2. X-ray diffraction pattern of the CFA raw material.



Figure 3. Relative abundance of dominant minerals in the CFA sample.

To better understand how the other relatively abundant minerals present in the CFA may possibly interfere with the extraction of the REEs of interest, it is important to know the association of the REEs with these dominant minerals. Figure 4 shows the SEM-EDS images with elemental map at different locations. It can be seen that the elemental map of Dy is almost identical with that of Fe, indicating a close association of Dy with Fe. Other REEs appear to be finely disseminated Throughout the matrix and show some associations with Al and Ca but they tend to appear more associated with Fe.



Figure 4. SEM-EDS images at \times 5000 magnification with elemental map at different locations of the CFA sample.

Figure 5 shows a SEM-EDS image with elemental map and point analysis. The elemental map confirms the close association of Dy with Fe qualitatively and the point analysis quantifies the association and shows a general trend that the amount of Dy is directly proportional to the amount of Fe except at some points.



(a) Mapping images at one point of the sample



	Mass %						
Display name	Fe	Dy					
Spc_066	69.64 ± 0.31	2.15 ± 0.17					
Spc_067	$50.10\pm\!\!0.23$	1.49 ± 0.13					
Spc_069	17.61 ± 0.13	0.56 ± 0.09					
Spc_070	83.06 ± 0.43	1.77 ± 0.24					
Spc_071	81.59 ± 0.43	1.90 ± 0.24					
Spc_072	67.03 ±0.45	2.05 ± 0.28					
Spc_074	1.20 ± 0.04	nd					
Spc_075	12.52 ± 0.11	0.16 ± 0.08					
Spc 076	24.84 ± 0.15	0.55 ±0.09					

(**b**) Elemental analysis of the point being mapped in the sample

Figure 5. SEM-EDS image at ×2300 magnification with elemental map and point analysis.

3.2. Optimization of Leaching Parameters Using Central Compiste Design-Response Surface Methodolgy

The highest extractions of Dy, Nd, Er, Eu and Tb were obtained at 3 M HCl, 65 °C and a leaching time of 270 min (Table 4) while the lowest extraction occurred at the lowest acid concentration 0.3M where concentrations of Dy, Er, and Eu were no longer detected by ICP-MS. This implies that the acid concentration of HCl was too weak to leach out the REE in the CFA. Also, at this acid concentration, only calcium (Ca), among the major elements that existed in the CFA, was leached out (Table 5). This inability of HCl at 0.3 M concentration to extract the major elements except Ca can be futher elucidated through the thermodynamic principles using the formula $\Delta G_{rxn} = \Delta G_{products} - \Delta_{reactants}$. At T = 323 K, the ΔG corresponding to the following reactions are shown [44]:

$$Al_2O_3 + 6HCl = 2AlCl_3 + 3H_2O; \quad \Delta G_{323K} = 401.5 \text{ kJ/mol}$$
 (2)

$$CaO + 2HCl = CaCl_2 + H_2O; \quad \Delta G_{323K} = -120.1 \text{ kJ/mol}$$
 (3)

$$Fe_2O_3 + 6HCl = 2FeCl_3 + 3H_2O; \quad \Delta G_{323K} = 150.8 \text{ kJ/mol}$$
 (4)

$$SiO_2 + 4HCl = SiCl_4 + 2H_2O; \quad \Delta G_{323K} = 286.2 \text{ kJ/mol}$$
 (5)

Table 4. Percent extraction of Dy, Er, Eu, Nd and Tb corresponding to each combination of independent variables.

Indeper C	ndent Var Coded For	iable in m	% Dy E	xtracted	% Er Ev	ctracted	% Eu E	xtracted	% Nd E	xtracted	% Tb E	xtracted
X ₁	X2	X ₃	Trial 1	Trial 2								
1	35	90	47.5	47.5	45.8	53.4	58.7	58.7	43.2	45.5	61.7	64.3
3	35	90	56.1	51.8	5 3.4	53.4	58.7	58.7	55.2	50.7	72.0	66.9
1	65	90	51.8	51.8	53.4	53.4	58.7	58.7	48.5	47.7	66.9	66.9
1	35	270	51.8	51.8	53.4	53.4	58.7	58.7	47.7	47.7	66.9	66.9
3	65	90	69.1	69.1	68.7	68.7	73.4	73.4	66.4	64.9	84.9	82.3
3	35	270	60.4	60.4	61.1	61.1	73.4	73.4	58.2	57.4	74.6	74.6
1	65	270	51.8	56.1	53.4	53.4	58.7	58.7	49.2	50.7	66.9	72.0
3	65	270	73.4	73.4	76.3	76.3	88.0	88.0	69.3	70.1	90.0	90.0
0.3	50	180	0.00	0.00	0.00	0.00	0.00	0.00	0.75	0.75	12.9	12.9
3.7	50	180	64.8	64.8	68.7	68.7	73.4	73.4	64.1	61.9	82.3	79.7
2	25	180	56.1	51.8	61.1	53.4	58.7	58.7	54.4	50.0	69.4	64.3
2	75	180	73.4	73.4	76.3	76.3	88.0	88.0	67.8	70.1	90.0	90.0
2	50	29	51.8	51.8	53.4	53.4	58.7	58.7	50.0	52.9	66.9	69.4
2	50	331	64.8	64.8	68.7	68.7	73.4	73.4	61.9	61.9	79.7	79.7
2	50	180	60.4	60.4	61.1	61.1	73.4	73.4	59.6	59.6	77.2	77.2
2	50	180	64.8	60.4	68.7	61.1	73.4	73.4	61.1	56.7	77.2	74.6
2	50	180	60.4	60.4	61.1	61.1	73.4	73.4	58.9	59.6	77.2	77.2
2	50	180	60.4	64.8	61.1	68.7	73.4	73.4	57.4	61.1	77.2	79.7

 $X_1 = acid concentration (M); X_2 = leaching temperature (°C); X_3 = leaching time (mins).$

Table 5. Percent extraction of Al, Ca, Fe and Si corresponding to each combination of independent variables.

Independent Variable in Coded Form			% Al Extracted	% Ca Extracted	% Fe Extracted	% Si Extracted
X ₁	X ₂	X ₃	Trial 1	Trial 1	Trial 1	Trial 1
1	35	90	21.2	78.2	13.2	20.3
3	35	90	26.0	79.9	21.1	18.8
1	65	90	24.4	81.6	11.9	20.3
1	35	270	22.8	78.2	13.2	20.3
3	65	90	37.4	79.9	48.8	20.3
3	35	270	29.3	79.9	27.7	18.8
1	65	270	24.4	83.3	6.59	20.3
3	65	270	42.3	83.3	59.3	18.8
0.3	50	180	0.00	49.3	0.00	0.00
3.7	50	180	34.2	81.6	43.5	6.2
2	25	180	24.4	79.9	18.5	21.9
2	75	180	39.0	85.0	47.5	23.4
2	50	29	24.4	78.2	19.8	20.3
2	50	331	32.5	81.6	33.0	20.3
2	50	180	30.91	81.62	29.01	21.88
2	50	180	30.91	81.62	29.01	21.88
2	50	180	29.28	81.62	26.37	21.88
2	50	180	29.28	79.92	26.37	21.88

 $\overline{X_1 = acid concentration (M)}$; $X_2 = leaching temperature (^{\circ}C)$; $X_3 = leaching time (mins)$.

Among the four reactions, only the leaching reaction of calcium oxide (Equation (3)) has a negative ΔG , which means that it is the only spontaneous reaction at 0.3 M HCl concentration. The failure of the 0.3 M HCl concentration to leach out the REEs can

be explained through the application of material balance where 5g of coal fly ash was leached with 50 mL of xM HCl. At very low HCl concentration, almost all of the H^+ ions were consumed by CaO which leaves a very small amount for the reaction with REE. Consequently, since no data was obtained at 0.3 M HCl concentration, it was excluded from the experimental design as including it would negatively affect the RSM analysis (ANOVA, diagnostics, Model graph, and optimization) in the Design Expert version 11 software.

To ensure a good model fitting, it was necessary to test for the significance of the regression model and its individual model coefficients. The significance of the factors were determined through the ANOVA where a factor or interaction between factors was considered significant if the *p*-value was less than 0.05. Otherwise, it was insignificant and thus excluded from the empirical model.

The results of the ANOVA for the quadratic model (Equation (1)) are shown in Tables 6–10 for Dy, Er, Eu, Nd and Tb, respectively. As can be seen in the tables, the three main factors, namely, acid concentration, leaching temperature and leaching time, have significant effects on the extraction of Dy, Er, Eu, Nd and Tb. The interactions between acid concentration and leaching temperature (AB) for all the critical REEs investigated were significant while all interactions involving leaching time (AC and BC) were insignificant except for the case of Eu (Table 8) where the interaction between acid concentration and leaching time (AC) was significant.

Table 6. ANOVA for Dy Extraction.

Source	Sum of Squares	df	Mean Square	F-Value	<i>p</i> -Value	
Model	1888.76	9	209.86	55.21	< 0.0001	significant
A-Conc	778.17	1	778.17	204.72	< 0.0001	significant
B-Temp	661.34	1	661.34	173.98	< 0.0001	significant
C-Time	223.26	1	223.26	58.73	< 0.0001	significant
AB	116.45	1	116.45	30.64	< 0.0001	significant
AC	4.66	1	4.66	1.23	0.2793	insignificant
BC	4.66	1	4.66	1.23	0.2793	insignificant
A ²	139.97	1	139.97	36.82	< 0.0001	significant
B^2	10.26	1	10.26	2.70	0.1135	insignificant
C ²	38.11	1	38.11	10.03	0.0042	significant
Residual	91.23	24	3.80			Ũ
Lack of Fit	35.33	4	8.83	3.16	0.0363	significant
Pure Error	55.90	20	2.79			U U
Cor Total	1979.99	33				

Table 7. ANOVA for Er Extraction.

Source	Sum of Squares	df	Mean Square	F-Value	<i>p</i> -Value	Remarks
Model	2079.34	9	231.04	22.66	< 0.0001	significant
A-Conc	823.65	1	823.65	80.79	< 0.0001	significant
B-Temp	646.58	1	646.58	63.42	< 0.0001	significant
C-Time	293.40	1	293.40	28.78	< 0.0001	significant
AB	178.46	1	178.46	17.51	0.0003	significant
AC	32.78	1	32.78	3.22	0.0856	insignificant
BC	3.64	1	3.64	0.3572	0.5556	insignificant
A^2	144.20	1	144.20	14.14	0.0010	significant
B^2	16.07	1	16.07	1.58	0.2214	insignificant
C ²	35.29	1	35.29	3.46	0.0751	insignificant
Residual	244.67	24	10.19			0
Lack of Fit	98.99	4	24.75	3.40	0.0283	significant
Pure Error	145.68	20	7.28			Ŭ
Cor Total	2324.01	33				

Source	Sum of Squares	df	Mean Square	F-Value	<i>p</i> -Value	
Model	3067.51	9	340.83	47.21	< 0.0001	significant
A-Conc	979.76	1	979.76	135.72	< 0.0001	significant
B-Temp	906.66	1	906.66	125.60	< 0.0001	significant
C-Time	427.22	1	427.22	59.18	< 0.0001	significant
AB	215.21	1	215.21	29.81	< 0.0001	significant
AC	215.21	1	215.21	29.81	< 0.0001	significant
BC	0.0000	1	0.0000	0.0000	1.0000	insignificant
A ²	319.35	1	319.35	44.24	< 0.0001	significant
B^2	0.0911	1	0.0911	0.0126	0.9115	insignificant
C ²	170.26	1	170.26	23.59	< 0.0001	significant
Residual	173.25	24	7.22			Ū
Lack of Fit	173.25	4	43.31			
Pure Error	0.0000	20	0.0000			
Cor Total	3240.76	33				

Table 8. ANOVA for Eu Extraction.

Table 9. ANOVA for Nd Extraction.

Source	Sum of Squares	df	Mean Square	F-Value	<i>p</i> -Value	Remarks
Model	1779.43	9	197.71	56.64	< 0.0001	significant
A-Conc	938.95	1	938.95	268.98	< 0.0001	significant
B-Temp	505.94	1	505.94	144.94	< 0.0001	significant
C-Time	147.33	1	147.33	42.21	< 0.0001	significant
AB	86.85	1	86.85	24.88	< 0.0001	significant
AC	3.47	1	3.47	0.9951	0.3284	not significant
BC	1.25	1	1.25	0.3583	0.5551	not significant
A^2	185.22	1	185.22	53.06	< 0.0001	significant
B^2	1.24	1	1.24	0.3550	0.5569	not significant
C^2	32.36	1	32.36	9.27	0.0056	significant
Residual	83.78	24	3.49			0
Lack of Fit	30.98	4	7.74	2.93	0.0465	significant
Pure Error	52.80	20	2.64			0
Cor Total	1863.21	33				

Table 10. ANOVA for Tb Extraction.

Source	Sum of Squares	df	Mean Square	F-Value	<i>p</i> -Value	
Model	2050.33	9	227.81	45.61	< 0.0001	significant
A-Conc	782.00	1	782.00	156.56	< 0.0001	significant
B-Temp	822.27	1	822.27	164.62	< 0.0001	significant
C-Time	205.57	1	205.57	41.15	< 0.0001	significant
AB	133.94	1	133.94	26.81	< 0.0001	significant
AC	6.61	1	6.61	1.32	0.2612	insignificant
BC	0.0000	1	0.0000	0.0000	1.0000	insignificant
A^2	130.63	1	130.63	26.15	< 0.0001	significant
B ²	1.99	1	1.99	0.3982	0.5340	insignificant
C^2	41.10	1	41.10	8.23	0.0085	significant
Residual	119.88	24	5.00			Ũ
Lack of Fit	53.74	4	13.43	4.06	0.0143	significant
Pure Error	66.14	20	3.31			
Cor Total	2170.21	33				

Table 11 shows the final empirical model which includes only the terms that were found to be significant. The negative values indicate antagonistic effects, whereas the positive values indicate synergistic effects [39].

%Extraction =		X ₁ (Conc)	X ₂ (Temp)	X ₃ (Time)	<i>X</i> ₁ <i>X</i> ₂	<i>X</i> ₁ <i>X</i> ₃	X1 ²	X ₃ ²
Nd	+59.3	+7.25	+4.3	+2.32	+2.33		-3.44	-1.16
Er	+63.1	+6.79	+4.87	+3.28	+3.34		-3.04	
Eu	+73.4	+7.40	+5.76	+3.95	+3.67	+3.67	-4.52	-2.65
Tb	+77.2	+6.61	+5.49	+2.74	+2.89		-2.89	-1.3
Dy	+61.5	+6.60	+4.92	+2.86	+2.70		-2.99	-1.26

Table 11. Final % Extraction Equation in terms of coded factors (eliminating insignificant terms; *p*-value > 0.05).

The magnitude of significance of the factors on the extraction of Dy, Er, Eu, Nd and Tb was measured through the Pareto chart of the standardized effects and is presented in Figure 6. It can be seen that the greatest effect on the extraction of the REEs of interest was contributed by the acid concentration, except for Nd where the effect of leaching temperature was more significant than the acid concertration.



Figure 6. Pareto charts of the standardized effects for individual critical REEs.

The adequacy of the developed model plays an important role in validating the analysis of the experimental data. Thus, fit statistics and diagnostics of the model were done and the results are summarized in Table 12. The predicted R^2 is a measure of how well the model predicts a response variable while the adjusted R^2 is used to determine the reliability of the correlation and how much is determined by the addition of independent variables. A good model should have a difference between predicted R^2 and adjusted R^2 not exceeding 0.2. Based on the fit statistics in the table, all models for the extraction of the REEs of interest qualify for a good model.

Model	Std. Dev.	Mean	C.V. %	R ²	Adjusted R ²	Predicted R ²	Adeq Precision
Dy Extraction	1.95	59.8	3.26	0.9539	0.9366	0.9079	27.1937
Er Extraction	3.19	61.52	5.19	0.8947	0.8552	0.7941	17.2442
Eu Extraction	2.69	69.03	3.89	0.9465	0.9265	0.8786	23.4966
Nd Extraction	1.87	57.1	3.27	0.955	0.9382	0.9073	27.3827
Tb Extraction	2.23	75.19	2.97	0.9448	0.924	0.8784	24.4924

Table 12. Fit statistics for the Extraction Models of critical REEs.

Another value that should be considered is the adequate precision (Adeq Precision), which compares the range of the predicted values at the design points to the average prediction error. It is used to measure the signal to noise ratio and a ratio of greater than 4 is desirable. According to Table 12, the Adeq Precision is greater than 17 for all extraction models indicating an adequate signal. Thus, the models can be used to navigate the design space.

Diagnostic plots such as the normal probability plots and the plots for predicted vs actual are shown in Figure 7. These plots are based on Externally Studentized Residuals method rather than on the Internally Studentized Residuals, as the former is more sensitive for finding the problems with the analysis compared to the latter. The Studentizing residuals map all the different normal distributions to a single standard normal distribution.

Figure 7a–e follow a straight line, which indicates that the data set follows a normal probability and thus there is no need for data transformations like logarithmic, square root or arc sine square root transformations. Moreover, Figure 7f–j showed straight lines for the predicted vs actual plots which means that the data set can be easily predicted by the model.

3.3. Optimization of Factors by RSM

The most important part of this study is the determination of the optimum leaching process condition where maximum critical REE extraction can be obtained from CFA. The optimum leaching condition was generated by the Design Expert Software for the REEs of interest and was found to fall at 3 M HCl concentration, 65 °C and 270 min of leaching time, which are the highest levels for each factor. There was a consistent increase in the REE extractions as the levels of the factors were increased within the chosen range for each factor. While the optimum extraction is expected as the levels of the factors are increased, this may not be necessarily true if precipitation of the target REEs occur. In the case of this study, a significant decrease in the extraction was not observed as the levels of the factors were increased, indicating that leaching with HCl at the chosen range of concentration, temperature and time did not lead to any precipitation and the optimum condition was successfully achieved.

The accuracy of the empirical models was tested by comparing the experimental and the predicted values at the optimum conditions; the results are summarized in Table 13. To be able to say that the model is accurate, the % error (i.e., the deviation of the predicted value from the experimental value) must be less than 5%. Based on the results in Table 13, all models showed errors less than 5%, therefore it is safe to say that the models established predicted the extraction of the REEs accurately.

Table 13. Comparisons between experimental and predicted results at the optimum level.

Critical REE	Experimental	Predicted	% Error
Dy	73.38%	75.01%	2.22%
Er	76.34%	78.89%	3.34%
Eu	88.02%	90.58%	2.91%
Nd	70.08%	71.34%	1.80%
Tb	90.01%	91.66%	1.83%



Figure 7. Diagnostic plots.

4. Conclusions

This research has proven that the Philippine CFA is a potential secondary resource of REEs based on the criteria of Seredin and Dai (2012) with an outlook coefficient of 1.03 and a relative percentage of critical elements of 33.3%. The REEs were found to be finely disseminated throught the matrix of the CFA sample; that is, they are associated with the other minerals present that are relatively abundant such as the aluminosilicates, oxides of iron and calcium. Strong associations of the REEs with Fe, espcieally Dy, was observed which indicates that Fe will be extracted together with the REEs. The statistical analyses conducted have shown that the three main factors (HCl concentration, leaching temperature and leaching time) investigated have significant effects on the extraction of Dy, Er, Eu, Nd and Tb, with HCl concentration as the factor that has the greatest effect on the response variable. The emperical models established through CCD-RSM for the extraction of the REEs of interest were found to be accurate with a prediction error of less than 5%. The optimum leaching conditions obtained using the RSM were 3M HCl concentration, 65 °C and 270 min with an extraction of 73.38%, 76.34%, 88.02%, 70.08% and 90.01% for Dy, Er, Eu, Nd and Tb, respectively.

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