

Secondary resources at abandoned mine tailings, Giyani Greenstone Belt, Limpopo province of South Africa

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Abstract

Mining waste are materials that result from the exploration, mining and processing of substances and can consist of natural materials, processed to varying degrees during the ore-processing and enrichment phases, and possibly containing chemical, inorganic and organic additives. Evaluation of the REEs potential of the Klein Letaba tailings dam was envisaged to assist in the containment at the currently un-rehabilitated mine tailings material. These tailings are a concern to the livelihood of nearby communities in the Giyani area due to their threat to water and soil quality. The methodologies used in this study consist of sampling, sample preparation and analysis using X-ray spectrometry technique and calculation of the tonnage of the tailings material using the Trapezoidal rule method. The results indicated that the REE abundance at the tailings dump in their decreasing concentration were $Yb > La > Ce > Gd > Sm > Dy > Y > Er > Tb > Eu > Sc$. The tailings dam was found to have significant levels of both LREEs and HREEs, which are above the upper continental crust thresholds. For instance, the HREEs Gd, Dy, and Yb were almost 10 times or more than the crustal abundances and this being 56, 46 and 17 mg/kg respectively. The LREEs and HREEs had also a total concentration 602 and 769 mg/kg respectively. The calculated volume of the tailings material was found to be 527,081 m³ and the tonnage was derived to be 1,291 349.0 tons of tailings material at the site. Consequently, the total LREEs and HREEs contained in the material was estimated to be 843 tons and worth re-evaluating. The paper uses REEs at this study site to illustrate how other un-rehabilitated sites can be re-evaluated and attract investment towards exploiting them and subsequently assisting in tailings containment. However, the REEs at Klein Letaba tailings dam was found to be uneconomic to exploit under the current prices and technology.

Keywords: *Rare earth elements, re-evaluation, rehabilitation, Giyani greenstone belt*

1 Introduction

The Giyani Greenstone Belt (GGB) situated in the north-eastern part of the Limpopo Province in South Africa is home to abandoned mine sites. These mine sites are a constant danger to nearby communities due to the risks associated with tailings dam, disused mine shafts, pit lakes and dilapidated mine buildings (Mhlongo et al., 2020). According to Steenkamp and Clark-Mostert (2012), the five largest of the inactive mines which operated in the belt include Franke, Birthday, Louis Moore, Klein Letaba and Fumani. Communities in the vicinity of various tailing dams in the GGB are under threat due to a great potential for problems arising from the un-rehabilitated gold mine tailings such as elevated dust levels, acid mine drainage, landslides, slumps and downward leaching of metals and processing chemicals, which may subsequently contaminate soil and underground water sources. Matovheke and Muzerengi (2018) reported elevated levels of Pb, Zn, Cd and As in the agricultural soils surrounding the Klein Letaba gold mine tailings dam. Elevated levels of Pb, Cr, Cd, As and Zn in soils in the vicinity of the abandoned Mandonsi mine in the belt have also been reported by Mulugisi et al. (2009) and Mitileni et al. (2011).

Mine tailings have been reported to be a cause of plumes of dust, which may cause asthma, chronic bronchitis, chronic cough, emphysema and pneumonia to nearby communities (Beaudry, 2018). Although many studies have focused on a multitude of metals contained in the GGB tailings material, studies have overlooked other metals such as the rare earth elements (REEs). The mobility and possible impact of REEs on

ecosystems are still relatively unknown and thus, potential risks to the environment and human health cannot be currently predicated. Elsewhere, study of the California gold tailings has reported gold mine tailings as mother lode of REEs (Farnahm, 2013). If the REEs could be extracted from previously mined ore, the environmental hazards of new mines and mining waste can be mitigated. According to Hoatson et al. (2011), REEs in Australia are associated with igneous, sedimentary, and metamorphic rocks from a wide range of geological settings.

The REEs are vital to some of the world's fastest growing markets for catalysts, magnets, ceramics, medical equipment, many defence applications and electronics. They underpin technologies that are critical for clean energy, transport and communication (Haque et al., 2014). There are fewer than twenty large companies that trade REEs, and these are distributed to a handful of countries such as China, Canada, Australia, USA, Russia, India and Japan. China is the biggest supplier of REEs in the world, while United States, Japan and Germany are the biggest importers (Roskill, 2015). About 95 % of all the world's REEs resources occur in three minerals, which are bastnasite ($\text{CeCO}_3(\text{OH}, \text{F})$), monazite ($\text{Ce, La} \text{ PO}_4$) and xenotime (YPO_4).

At Klein Letaba mine, huge piles of tailings were improperly disposed and left on the surface after closure of the mine around the 1990s. The numerous negative impacts associated with mine tailings near communities and coupled with huge rehabilitation cost presents a challenge to the local and national authorities. Consequently, any move towards exploiting these tailings material is a plausible initiative which may assist in the containment of the tailings material. This study seeks to investigate REEs concentrations at the Klein Letaba gold mine tailings dam. The possible recovery of REEs from the huge mine tailing dams is an exciting novel initiative for the area representing a dramatic paradigm shift for exploiting mine waste and consequently assisting in rehabilitating mine dump sites.

2 Study Area

The Klein Letaba gold mine tailings dam is situated on the south-western portion of the GGB about 20 km south-west of the present-day town of Giyani and roughly 85 km west of the Kruger national park. It is located between the latitudes $23^{\circ}17'38''\text{S}$ and $23^{\circ}17'48''\text{S}$ and longitudes $30^{\circ}33'07''\text{E}$ and $30^{\circ}36'25''\text{E}$ (Figure. 1).

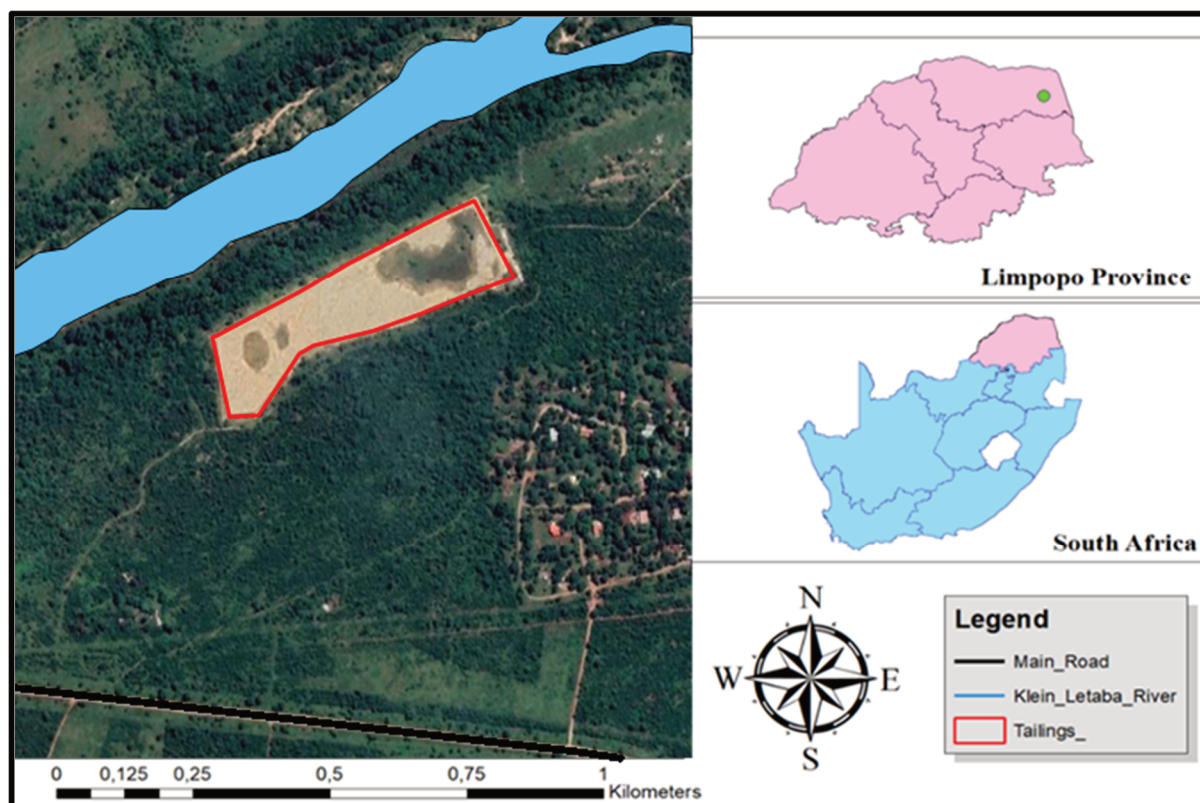


Figure 1 A map showing the location of the Klein Letaba gold mine tailings dam

2.1 Geological setting

The GGB which trends from south-west towards the north-east is well known for its gold mineralisation and it occupies the north-eastern edge of the Kaapvaal Craton. It has got an approximate length and width of 70 and 15 km respectively. The belt is of Archean age and the lithological succession has been designated as the Giyani Group of the Swazian Supergroup (SACS, 1980). Generally, the GGB is poorly exposed hence there has been no attempt to correlated various units across the belt or to establish stratigraphic units (McCourt and Reenen, 1992). According to Brandl et al. (2006), the GGB appears to be dominated by mafic rocks, ultramafic and minimal felsic schists, iron-formations and meta-sediments. Quartz-biotite, garnetiferous Fe-rich quartz-amphibole schists, and quartz-muscovite are the minor metasedimentary rocks in the belt (Pretorius et al., 1988). On its southern and western part, the GGB bifurcates into two narrower arms of 3-5 km width with an outcrop length more than 50 km (McCourt and Van Reenen, 1992). Prinsloo (1977) has termed these linear greenstone assemblages, the Khavagari arm (in the north) and Lwaji arm (in the south). The northern part of the belt is composed mainly of tremolite-actinolite schist, amphibolites, and banded iron formation, while the southern part consists mainly of amphibolites, banded iron formation, tremolite-actinolite schist, and quartz-sericite schist (Gan and Van Reenen, 1995).

An envelope of migmatitic gneisses surrounds the GGB (McCourt and Reenen, 1992) and a few localities where contacts are exposed tectonically interleaved greenstones and gneisses can be seen (Brandl et al., 2006). According to Potgieter and de Villiers (1986), the auriferous unit contains the banded iron-formation, amphibolites, and tremolite-actinolite schist. Geophysical modelling undertaken by Kleywegt et al. (1987) indicated a shallow structure of about 1.5 km depth with maximum depth of 3 km towards its SE margin of the belt. Micaceous quartzite, iron formation and a spotted ultramafic rock occupy the far northeast region of the belt (Pretorius, 1988).

3.1 Sampling and analysis

A total of 20 samples were collected from the tailings dam along five lines oriented approximately north-south with 25 metres as sampling interval using a hand auger and samples were subsequently transported to the laboratory for analysis. Sampling implements and other work surfaces were thoroughly cleaned between samples during sampling, preparations and analysis. A riffle splitter was used to split each of the 20 samples to smaller portions for drying and milling using a Vacutec drying Oven and Retsch, RS 200 milling machine.

The samples were then prepared as pressed powder pellets with a binding agent, boric acid. Approximately 10 g of the milled powder were pressed in aluminium cups (40 mm in diameter) on a bed of 2.5 g boric acid while applying a pressure of 30-40 tons for 20 seconds using a hand operated press. An S2 Ranger XRF instrument was employed to analyse the REEs and a set of 6 reference standards were used to set up calibration for the instrument. The analyses were carried out in duplicate for the 20 samples for low rare earth elements (LREEs) La, Ce, Sm, Eu, Sc, and high rare earth elements (HREEs) were Yb, Gd, Dy, Y, Er and Tb.

3.2 Determining REE tonnage at Klein Letaba

The authors employed the Trapezoidal Rule for area and volumes method in order to obtain an estimate of the tonnage of the total tailings contained at Klein Letaba. According to Schofiel and Breach (2007), and Chandra (2005), areas and volumes of enclosed by irregular boundaries like the mine tailings or stockpiles can be accurately estimated by the Trapezoidal rule method. In order to calculate total tonnage of the tailings material, the area was divided into a number of trapezoids with the boundaries assumed to be straight at the adjacent offsets. Consequently, the tailings dam was subdivided into two segments (segment A and B) comprising of 9 and 8 trapezoids with total lengths of 120.0 and 330.0 m respectively (Figure 2). The two segments were made up of 10 and 8 offsets of varying length and had uniform width among them of 15.0 and 33.0 m respectively. By using this method, area of each trapezoid was determined and combined to derive the total area. The following formula was used:

$$A = d \left(\frac{O_1 + O_n}{2} + O_2 + O_3 + \dots + O_{n-1} \right) \dots \dots \dots (i)$$

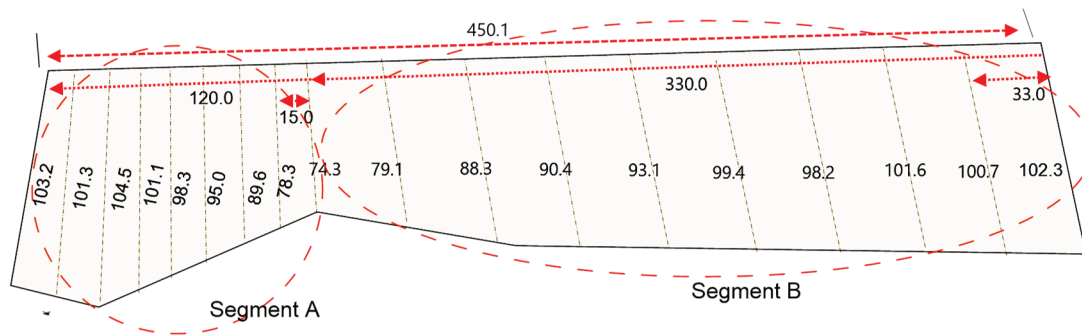


Figure 2 A Sketch of how the tailings dam was demarcated for area and volume calculation.

Using the equation above, the total area for the segments were found to be 39043 m². In order to obtain the total volume of the material, the height of the tailings dam was measured and found to be 13.5 m. Therefore, using area obtained above, the volume of the tailings material maybe estimated as follows;

Volume of earth work, $V = A \times L$, where L is equivalent to the height obtained and being 13.5 m. Therefore, the total volume was found to be 527,081m³ (39,043 m² x 13.5 m). Subsequently, the volume obtained was converted to tonnage and this was done by measuring the density of the tailings material which was found to be 2.45 g/cm³. Using this data, the tonnage of tailings was calculated as:

$$\text{Density} = \frac{\text{Mass}}{\text{Volume}} \dots \dots \dots (ii)$$

Tonnage of material = 1,291,349 tons (Density x tailings volume).

4 Results and Discussion

4.1 Statistical analysis

Summary statistics of geochemical data of the analysed REEs are shown below (Table 1). REEs in their order of decreasing mean concentration were: Yb > La > Ce > Gd > Sm > Dy > Y > Er > Tb > Eu > Sc. The other REEs such as Ho, Tm, Pr, Nd, Pm and Sc were below the instrument's detection limit of <1 ppm. The total REEs concentration ranged from 303 to 1371 mg/kg. HREEs concentration was much higher with a mean concentration of 390 mg/kg. Cerium and La had the highest concentration of 252 and 266 mg/kg respectively whilst, Sc was lowest with 0.5 mg/kg. The average total content of REEs (Σ REE) was 674 mg/kg with a range of 105 mg/kg.

The study used coefficient of variation (CV) to investigate the variability of the data. Low CV values correspond to a homogeneous distribution whilst, high CV values represents heterogenous distribution. The CVs of all REEs were relatively low with a minimum of 7 and consequently, indicating a homogeneous distribution of the REEs in the tailings material. However, the LREEs were less homogenous than the REEs with mean CV values of 37.7 and 19.4 respectively. The concentration range for most of the REEs were low and therefore indicating low variability within the tailings material. In general, HREEs are more likely to have more mobility than LREEs and thus, LREEs are enriched and HREEs are depleted gradually during the migration processes as reported by Masto et al. (2011). According to Edahbi (2017), REEs are associated with silicates, carbonates, fluo-carbonates, oxides, phosphates, and sulfates bearing lithologies and these are prevalent in the Giyani Greenstone belt. The LREE/HREE ratio ranged largely from 0.61 to 0.84 with an average of 0.73. The LREE/HREE ratio showed that the content of LREE does not vary significantly than that of HREE in the Klein Letaba mine tailings. The spatial distribution of REEs is useful in assessing the possible sources of REEs enrichment and also to identify areas with anomalous REEs concentration. Apart from Sc, Y and Ce, the mean REEs concentration in the tailings dam was two or more times higher than the UCC averages.

Table 1 Statistical data of the analysed REEs at Klein Letaba tailings dam (max and min concentration in mg/kg)

REE	Min	Max	Range	Mean	SD	%CV	UCC
Sc	0.4	0.5	0.1	0.5	0.05	11.4	7
Y	7.2	11.5	4.3	8.7	1.1	12.9	21
La	98.5	266.1	167.6	167.0	40.04	24.0	30
Ce	5.2	252.1	246.9	62.3	65.00	104.3	64
Sm	9.4	81.2	71.8	53.2	18.23	34.3	4.5
Eu	1.5	2.0	0.5	1.8	0.13	7.0	0.88
Gd	35.8	70.7	34.9	55.9	9.45	16.9	3.8
Tb	2.5	5.3	2.8	4.2	0.84	20.2	0.64
Dy	29.1	58.4	29.3	45.5	8.77	19.3	3.5
Er	4.5	8.9	4.4	7.2	1.34	18.8	2.3
Yb	6.77	38.1	17.6	16.7	11.3	28.2	2.2
ΣREE	303.1	1370.7	31.3	674.3	327.5		-
LREE	115.0	601.9	486.9	284.8	123.4		-
HREE	188.1	768.8	580.7	389.5	204.1		-
LREE/HREE	0.61	0.84	0.23	0.73		-	-
SD, standard deviation; CV, coefficient of variation; UCC, average upper continental crust (UCC from Castor and Hedrick (2006))							

The P-values between cross pairs of REEs varied from <0.05 to >0.05 implying that correlation between the cross pairs varied from being statistically significant to insignificant (Table 2). Strong correlation coefficient was observed among pairs such as La-Y, Y-Ce, Sc-Tb and Sc-Er with correlation coefficients of 0.78, 0.86, 0.82 and 0.77 respectively. The HREEs pairs had P-values < 0.05 implying a statistically significant correlation between them. Strong correlation between REEs pairs could be attributed to the REEs emanating from the same source.

Table 2 Pearson's correlations between REE concentrations in the surface soil samples (p < 0.05)

	La	Ce	Sm	Eu	Sc	Gd	Tb	Dy	Er	Y
Ce	0.373									
Sm	<0.001	0.112								
Eu	0.027	0.413	<0.001							
Sc	0.671	0.681	0.451	0.226						
Gd	<0.001	0.095	<0.001	<0.001	0.500					
Tb	<0.001	0.112	<0.001	<0.001	0.818	<0.001				
Dy	<0.001	0.160	<0.001	<0.001	0.675	<0.001	<0.001			
Er	<0.001	0.110	<0.001	<0.001	0.774	<0.001	<0.001	<0.001		
Y	0.783	0.861	0.450	0.169	0.440	0.311	0.402	0.373	0.415	
Yb	<0.001	0.129	<0.001	0.001	0.571	<0.001	<0.001	<0.001	<0.001	0.804

4.2 LREE Concentration

LREEs in their order of decreasing concentration values were $\text{La} > \text{Ce} > \text{Sm} > \text{Eu} > \text{Sc}$ and their concentration ranged from 0.4 to 266 mg/kg (Figure 3). Ce and La had the highest concentration of 252 and 266 mg/kg respectively whilst Sc and Eu indicated maximum concentration of 0.5 and 2.0 mg/kg.

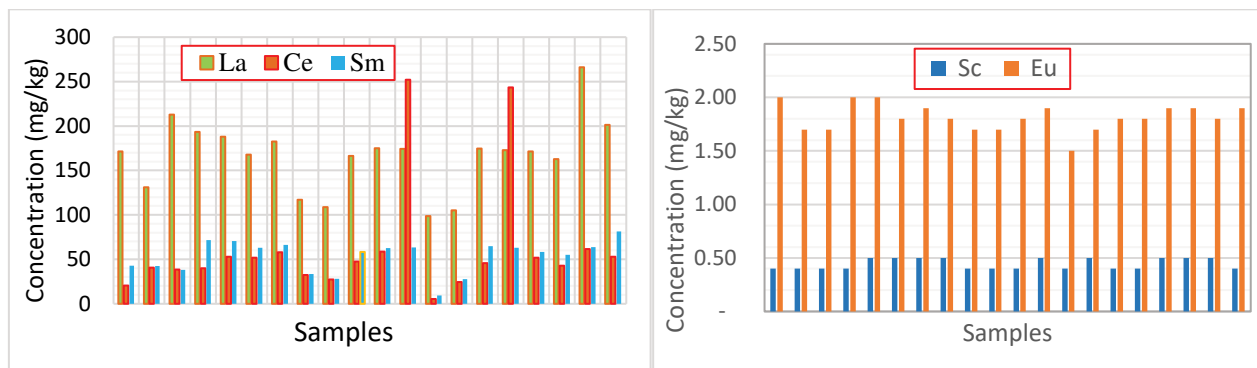


Figure 3 LREE concentration in Klein Letaba gold mine tailings

3.3 HREE Concentration

HREEs in their order of decreasing mean concentration values were $\text{Yb} > \text{Gd} > \text{Dy} > \text{Y} > \text{Er} > \text{Tb}$ with variable levels ranging from 2.5 to 71 mg/kg (Figure 4). The concentrations of the three HREEs (Gd, Dy, and Yb) at Klein Letaba tailings were almost 10 times or more than the crustal abundances and this being 56, 46 and 17 mg/kg respectively. The other Tb and Er were at least two times higher than the crustal abundances whilst, Y concentration was lower than the UCC averages. Considering that these REEs occur as residues of gold mining operations, the accumulation of REEs at the tailings dump is attributed to ore dressing processes which was employed at the mine site.

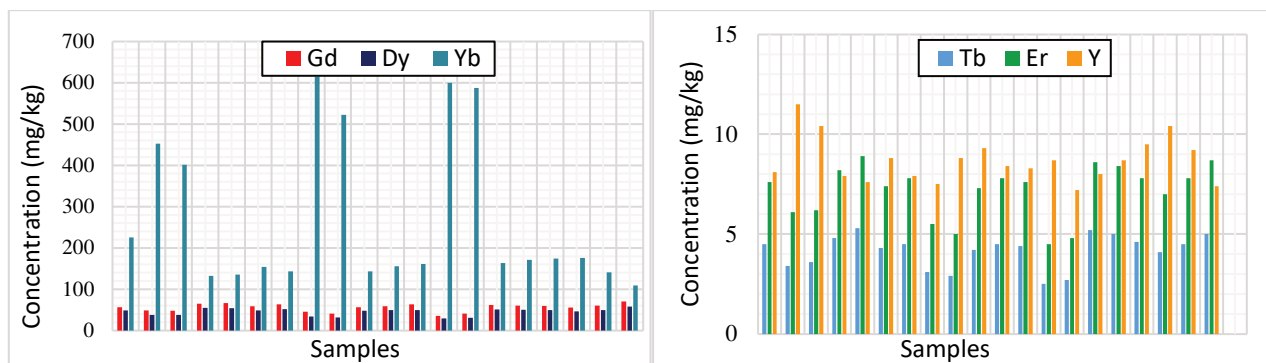


Figure 4 HREE concentration in Klein Letaba gold mine tailings

4.4 REE differentiation patterns

The REEs sample concentrations are usually normalized against the composition of the upper continental crust and in this study, normalization was undertaken using McDonough and Sun (1995) normalisation factors. Normalizing the REEs concentration to chondrites is a convenient way for eliminating the Oddo-Harkins effect and characterize the REEs signature in the Klein Letaba tailings material. The Oddo-Harkins rule states that the cosmic abundance of elements with an even atomic number is greater than that of adjacent elements with odd atomic number. Consequently, the plot of relative atomic abundance against increasing atomic number displays a ‘toothed’ curve, rather than a smooth line. It should also be noted that the REEs composition and differentiation patterns are not influenced by weathering processes and the REEs concentration remains unchanged. The REEs distribution patterns of the samples were high from the left through elements Ce, Sm, to Er suggesting that the LREE and HREE are fractionated in different degrees (Figure 5). The REEs exhibited negative Ce, Eu and Er anomaly with a ‘V’ shape at the REEs normalized

distribution patterns. With exception of Sc, Ce and Y, REEs concentration at Klein Letaba tailings dam were above the average upper continental crust described by Taylor and McLennan (1995) and similar to that of Humphris (1984) and Henderson (1984).

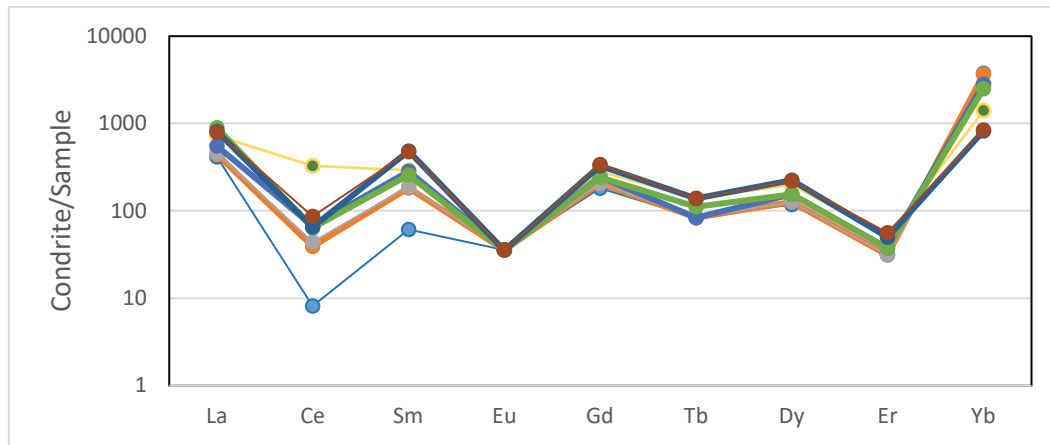


Figure 5 Chondrite normalized REE abundances of the samples collected from Klein Letaba gold mine tailings

4.5 REEs tonnage at Klein Letaba

The world has only a few mines that supply the entire world with REEs and most of them are located in China such as the largest Bayan Obo REEs mine (Humphries, 2013). Currently, REEs are mined very cheaply in China as a by-product of iron production at Bayan Obo mine, or from lateritic rich clays in southern China (Hoatson, 2011). The estimated weight of LREEs and HREEs contained in the tailings material may be derived from the average grade of these elements and this being 284.8 and 389.5 mg/kg. This converts to 368 and 475 tons of LREEs and HREEs at Klein Letaba tailings dam. The total tonnage of 843 tons of REEs at this tailings dump is quite significant. Although the REEs grades are not lucrative for exploitation under the current prices and with current technology, their levels are above the upper continental crust and worth evaluation in detail. Since the tailings material is finely ground (< 2 mm), this will also reduce beneficiation costs which may go a long way towards covering the rehabilitation cost of these mine tailings.

5 Conclusions

Studies by Salminen et al. (2005) stated that Ce and Sm are normally associated with felsic and mafic igneous rocks and some of these lithologies are also found within the Giyani Greenstone belt. Klein Letaba tailings dam has significant levels of both LREEs and HREEs, which are above the upper continental crust thresholds. According to Taylor and McLennan (1985), the upper crustal abundances of Gd, Dy, and Yb range from 2 to 4 mg/kg. The concentrations of these three HREEs at Klein Letaba tailings were almost 10 times or more than the crustal abundances and this being 56, 46 and 17 mg/kg respectively. The other Tb and Er were at least two times higher than the crustal abundances whilst, Y concentration was lower than the UCC averages.

The spatial variability of REEs concentration were low indicating a homogeneous distribution of the REEs in the tailings material. The REEs distribution patterns at Klein Letaba tailings dam as shown by Chondrite normalized REE pattern is in agreement with Wang and Liang (2015) and Laveuf, and Cornu (2009) who reported that individual REEs tend to decrease with increasing atomic number, and REEs with even atomic numbers are more frequent than their neighbours with odd atomic numbers. The LREEs/HREEs ratio indicated that the content of LREEs does not vary significantly with HREEs in the Klein Letaba mine tailings. Tailings dams have been a major environmental challenge in the GGB and all over the world and REEs in the tailings material could pollute communal lands in the vicinity of the tailings dam. The estimated tonnage of 843 tons of the REEs in the tailings material is quite significant. It may be worthwhile to establish the cutoff grade of the REEs whilst considering the available technologies to extract these metals from the Klein tailings dam.

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