

Review

Bio-Reclamation of Strategic and Energy Critical Metals from Secondary Resources

Sadia Ilyas ^{1,*}, Min-Seuk Kim ², Jae-Chun Lee ^{2,*}, Asma Jabeen ¹ and Haq Nawaz Bhatti ¹

¹ Mineral and Material Chemistry Laboratory, Department of Chemistry, University of Agriculture Faisalabad (UAF), Faisalabad 38040, Pakistan; asmajabeen7861@gmail.com (A.J.); profhnbhatti@gmail.com (H.N.B.)

² Mineral Resources Research Division, Korea Institute of Geoscience and Mineral Resources (KIGAM), Daejeon 305-350, Korea; redkms@kigam.re.kr

* Correspondence: sadiailyas1@yahoo.com (S.I.); jcllee@kigam.re.kr (J.-C.L.); Tel.: +92-41-9200161-69 (S.I.); +82-42-868-3612 (J.-C.L.)

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Abstract: Metals with an average crustal abundance of <0.01 ppm, which are high in supply shortage due to soaring demand, can, under the excessive environmental risk and <1% recycling rate of their production, be termed as ‘critical’ in a limited geo-boundary. A global trend to the green energy and low carbon technologies with geopolitical scenario is challenging for the sustainable reclamation of these metals from secondary resources. Among the available processes, bio-reclamation can be a sustainable technique for extracting and concentrating these metals. Therefore, in the present paper, the potential reclamation of critical metals (including rare earth elements, precious metals, and a common nuclear fuel element, uranium) via their interaction with microbe/s has been reviewed.

Keywords: bio-reclamation; energy critical metals; microbes-metal interaction; sustainability

1. Introduction

The rare earth elements (REEs) and precious metals (PMs) have remained in metallurgical attraction for a long time. The tendency to occur together in nature, difficulties in their extraction and separation with specific area of applications has placed these metals in the list of critical metals. However, in the recent past, the definition of criticality is certainly not limited to the difficult metallurgy and abundance in the earth’s crust (<0.01 ppm). In this technological era, the metals required in advanced materials and technologies for the transition towards a sustainable post-fossil society facing high supply risk due to the market complexities, lack of supply diversity, and geopolitical scenarios are termed as ‘critical metals’ [1]. Metals in the list of criticality changes with region or country. The organizations like, the American Physical Society and Materials Research Society have termed ‘energy-critical elements’ (ECEs) as the elements necessary for emerging renewable energy whose unavailability could limit future research & development (R&D) or deployment of a technology [2]. On the other hand, the European Union (EU) has accounted for the political and economic stability of producing countries along with their weak environmental performance to calculate the supply shortage of a particular metal [3]. Hence, commonly a metal can be labelled ‘critical’ which is facing high shortage in supply, excessive environmental risk and be of significant economic importance under the availability of a limited geo-boundary. In that case, the criticality of metal can simply be presented as follows [4]:

$$\text{Criticality} = \text{Supply risk} \times \text{Vulnerability} = \text{Likelihood of supply disruption} \times \text{Economic Consequences} \quad (1)$$

Notably, the availability of these metals largely influences the energy supply; country’s economy, export business, job creation and geopolitical relation of two individual countries [5], hence can also be

pronounced as ‘strategic critical metals’. Figure 1 presents the list of such metals, including uranium (the major source of non-fossil, nuclear fuel) and gold (widely used for plating other conductor materials besides its economic importance in global trade) [2,3,6]. In the scenario of a mineral deficit country like Korea, all the metals shown in Figure 1 can be on the country’s critical-list. Most of these metals are currently being exploited through mining and metallurgy of their fast depleting primary ores in their respective region. Hence, the sustainable and cost-effective processing of these metals from the waste/end-of-life/scrap materials as their alternative sources is an absolute necessity of time. Moreover, metal contamination from waste materials is a major concern because: metals are non(bio)degradable substances which can transform to form their carcinogenic and toxic compounds to transfer across the trophic levels of food-chain and bioaccumulating in living organisms [7–9]. Therefore, reclamation of critical metals from their alternative secondary resources can be a game-changer by overcoming the geo-boundary limitation, mitigating the supply risk and controlling the environmental damage by mining of their primary ores.

○ Rare Earth Element □ Energy Critical Element ▽ Other Critical Element

Scandium 21 Sc 44.96	Titanium 22 Ti 47.88	Vanadium 23 V 50.94	Chromium 24 Cr 52.00	Manganese 25 Mn 54.94	Iron 26 Fe 55.85	Cobalt 27 Co 58.93	Nickel 28 Ni 58.69	Copper 29 Cu 63.55	Zinc 30 Zn 65.39	Gallium 31 Ga 69.72	Germanium 32 Ge 72.61	Arsenic 33 As 74.92	Selenium 34 Se 78.96
Yttrium 39 Y 88.91	Zirconium 40 Zr 91.22	Niobium 41 Nb 92.91	Molybdenum 42 Mo 95.94	Technetium 43 Tc (98)	Ruthenium 44 Ru 101.07	Rhodium 45 Rh 102.91	Palladium 46 Pd 106.42	Silver 47 Ag 107.87	Cadmium 48 Cd 112.41	Indium 49 In 114.82	Tin 50 Sn 118.71	Antimony 51 Sb 121.76	Tellurium 52 Te 127.60
Lutetium 71 Lu 174.97	Hafnium 72 Hf 178.49	Tantalum 73 Ta 180.95	Tungsten 74 W 183.84	Rhenium 75 Re 186.21	Osmium 76 Os 190.23	Iridium 77 Ir 192.22	Platinum 78 Pt 195.06	Gold 79 Au 196.97	Mercury 80 Hg 200.59	Thallium 81 Tl 204.38	Lead 82 Pb 207.20	Bismuth 83 Bi 208.98	Polonium 84 Po (209)
Lanthanum 103 La (262)	Rutherfordium 104 Rf (267)	Dubnium 105 Db (268)	Seaborgium 106 Sg (271)	Bohrium 107 Bh (272)	Hassium 108 Hs (270)	Mtnerium 109 Mt (276)	Darmstadtium 110 Ds (281)	Roentgenium 111 Rg (280)	Copernicium 112 Cn (285)	Ununtrium 113 Uut (284)	Ununquadium 114 Uuq (289)	Ununpentium 115 Uup (288)	Ununhexium 116 Uuh (293)
Lanthanum 57 La 138.91	Cerium 58 Ce 140.12	Praseodymium 59 Pr 140.91	Neodymium 60 Nd 144.24	Promethium 61 Pm (145)	Samarium 62 Sm 150.36	Europium 63 Eu 151.97	Gadolinium 64 Gd 157.25	Terbium 65 Tb 158.93	Dysprosium 66 Dy 162.50	Holmium 67 Ho 164.93	Erbium 68 Er 167.26	Thulium 69 Tm 168.93	Ytterbium 70 Yb 173.04
Actinium 89 Ac (227)	Thorium 90 Th 232.04	Protactinium 91 Pa 231.04	Uranium 92 U 238.03	Neptunium 93 Np (237)	Plutonium 94 Pu (244)	Americium 95 Am (243)	Curium 96 Cm (247)	Berkelium 97 Bk (247)	Californium 98 Cf (251)	Einsteinium 99 Es (252)	Fermium 100 Fm (257)	Mendelevium 101 Md (258)	Nobelium 102 No (259)

Figure 1. Periodic table indicating the critical and scarce metals taken into this study [1–3].

A variety of reclamation routes have been explored and reviewed so far, including high calorific smelting, large chemical intensive hydrometallurgy, and energy intensive electrometallurgy [6,10–12]. However, the microbe-to-metal/material interactions have attracted renewed attention for a sustainable reclamation of metals from diversified materials [9]. Bio-reclamation including bacterial leaching, control of redox reaction using bacterial treatment, metal adsorption and (impurity) removal by employing dead biomass as the potential sorbents can facilitate the recycling of various critical and valuable metals from diverse secondary materials [13,14]. In this article, we have reviewed the literatures on bio-reclamation techniques (which includes microbial leaching, sorption, reduction and accumulation processes from non-primary sources) of critical metals; wherein the microbes-to-metal interaction, process sustainability and future challenge for the practical implementation of these processes has been discussed.

2. Potentially Important Microbes and Mechanism of Bio-Reclamation

From a bio-reclamation perspective, potentially important bacteria belong to the genus *Acidiphilium*, *Acidithiobacillus*, *Sulfobacillus* and *Leptospirillum* [15]. The most vital bio-reclaiming

mesophiles belong to *Acidithiobacillus*, including sulfur and iron oxidizing *Acidithiobacillus ferrooxidans* and sulfur oxidizing *Acidithiobacillus thiooxidans*. *Acidithiobacillus ferrooxidans* supplied with a broad metabolic capacity was the first acidophilic bacteria which was used for the bio-metallurgical perspectives. The investigations at the putative boundary of mesophilic and moderately thermophilic conditions at a temperature of 45 °C, showed *Acidithiobacillus caldus* being a dominant sulfur oxidizer; whereas *Leptospirillum* dominates as iron oxidizer. The role of Fe-oxidizing, *Leptospirillum ferrooxidans* in bio-reclamation was lately recognized due to its poor enrichment from samples as compared with the *Acidithiobacillus ferrooxidans*. The prior importance of *Leptospirillum ferrooxidans* in bio-reclamation was due to its mix culture with *Acidithiobacillus thiooxidans*, which could oxidize pyrite on a faster rate than using *Acidithiobacillus ferrooxidans* alone. Such behavior was ascribed to: its high propensity to attach with materials to be leached, a high affinity for Fe²⁺ and the low sensitivity to inhibition by Fe³⁺ [16,17].

Association between the microbial activity and exothermic reaction caused in sulfur containing materials points out the relevance of thermophiles in bio-reclamation process. Moderately thermophilic iron and sulfur-oxidizing microbes were primarily cultured from the mining conditions or hot springs [18,19] and subsequently from the mine-sites of iron ore [20]. Advancement in moderate thermophiles such as *Sulfobacillus thermosulfidooxidans* continues in recent mining operations, including refractory gold heaps undergoing biooxidation [21] and in bio-reclamation of electronic wastes [20,22]. Nowadays extremely thermophilic microorganisms in acidic, sulfidic hot springs have been used for their exceptional ability to oxidize the hard-to-treat materials. These microorganisms have enlarged the bio-reclamation options in terms of temperature and metal tolerance capacity. Figure 2 depicted various modes of microbial-metal interactions for metal mobilization or immobilization.

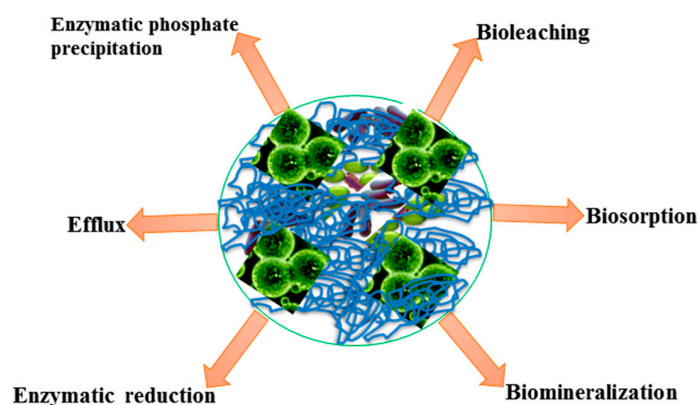
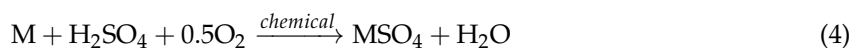
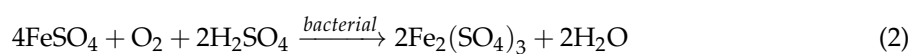


Figure 2. A typical representation of various modes of bio-reclamation process.

Mostly acidophilic autotrophs generate ferric iron and sulfuric acid for biooxidation and enhanced dissolution of associated metals (M) or metal oxide (MO) by direct, indirect, thiosulfate and poly-sulfide mechanisms after additional supplementation of sulfur source as in Equations (2)–(6) and Figure 3 [23,24]. Other than the acidophilic autotrophs, some heterotrophic microbes (*Pseudomonas fluorescens*, *Chromobacterium violaceum*) and fungi (*Penicillium simplicissimum*, *Aspergillus niger*) have also shown potential in bio-reclamation of metals. Employing the heterotrophic microorganisms, metals are usually reclaimed by displacement with H⁺ ions or formation of soluble metal species/chelates or the cyanidation with biogenic CN⁻ [25,26].



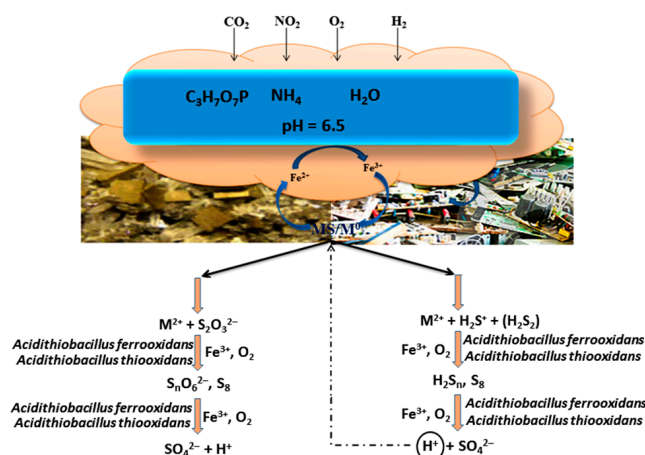
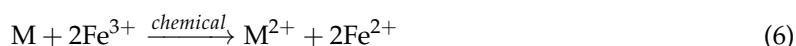
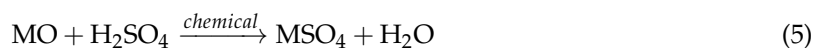


Figure 3. Interaction of *Acidithiobacillus ferrooxidans* with metals on surface of source material [23].

3. Bio-Reclamation of Carritical Metals

3.1. Bio-Reclamation of Rare Earth Elements

The rare earth elements (REEs) are a combination of 17 chemically similar elements, comprised of 15 lanthanides (Z, 57 to 71), scandium (Z, 21) and yttrium (Z, 39), and exhibiting the characteristics of magnetism, fluorescence, and superconductivity. REEs are being used in rechargeable batteries, permanent magnets, catalysts, lamp phosphors, and biomedical applications. Based on the importance, they are sub-grouped as light (La to Sm) and heavy (Eu to Lu) REEs. The heavy REEs are technologically vital for the electronics, defence, and low-carbon renewable energy, and identified as the most critical raw materials with an aggregate demand of ~210,000 tonnes by 2015 [27]. Figure 4 depicts the probable interaction of microbes with REEs and their subsequent leaching. However, exact understanding of the interactions is still not well documented despite a clear interest for developing biotechnology in clean recovery from the mine-sites. Consequently, only a few literatures have reported reclaiming REEs from secondary resources. A study involving the bio-reclamation of REEs from red mud describes the use of *Penicillium tricolor* at 2% (*w/v*) pulp density, could result in the maximum extraction of REEs and radioactive elements [28]. The maximum extraction was achieved under a two-step bio-reclamation process using 10% pulp density. This study provides a basis for seeking the bio-reclamation of REEs from other secondary resources, including the electronic wastes.

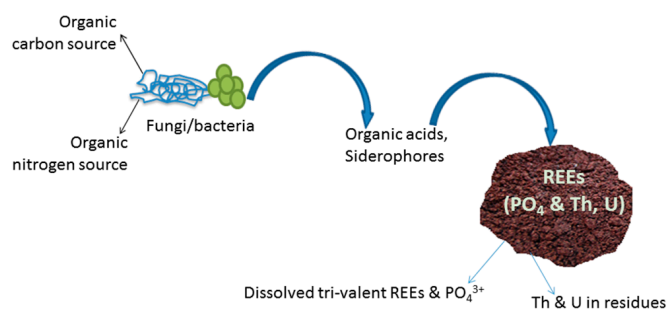


Figure 4. Bio-reclamation of rare earth elements by interactions with heterotrophic microbes.

The bioleaching in presence and absence of Fe/S oxidizing bacteria (*Acidithiobacillus ferrooxidans*) has been studied for yttrium extraction from the waste cathode ray tube (CRT) glass [29]. Notably, the effect of ferric ions can be understood by its ability to influence the redox potential of a bioleaching system [30]. The results revealed a linear enhancement in Y extraction with time, which could reach the maximum yield of 40% and 70% after a 16 days of bioleaching in the presence and absence of ferric ions, respectively [29]. The improved extraction of Y in the absence of ferric ions, attributed to the bacterial activity, was the highest with 3×10^8 cells/mL. A decrease in Y extraction with increase in pulp density indicated its negative effect on bacterial growth. The toxicity generated by the high metal content [31] in CRT (43% Zn and 23% Y) has been identified as the technological limit of this process. Hopfe et al. [32] recently reported the bioleaching of REEs from waste fluorescent powder (12.1% Y, 1.4% Ce, 0.8% Gd, 3.4% La, 0.7% Eu, 0.6% Tb) by using tea fungus Kombucha. The highest leaching yields were obtained in shake cultures either by using the entire Kombucha-consortium or with the supernatant of Kombucha, compared to the leaching yields of isolated *Zygosaccharomyces lentus* and *Komagataeibacter hansenii*. A primarily and effective leaching of Y and Eu indicated the red dye of $Y_2O_3:Eu^{3+}$ could be leached selectively. Approximately 60% and 68% Y was leached after 15 days of bioleaching in stationary and shaken mode, respectively; however, only 0.01% La could be released in both modes. Initially in both cases, a rapid decrease in pH was observed due to the in-situ generation of organic acids, mainly the acetic and gluconic acid. After a prolonged time (over 3 days) it slowed down due to the buffering effect of weak acids. Similar trends on pH profile with in-situ production of organic acids in the bioleaching of electronic waste, red mud, cracked catalysts and monazite ores have been observed in earlier studies [28,33–36]. Reed et al. [36] reported that 60% of the organic acids, mainly gluconic acid, can be produced within the initial 24 h. Using the isolated microbial culture BH1, *Acinetobacter calcoaceticus* GSN8; BH24, *Pseudomonas frederiksbergensis* 37; and A1, *Talaromyces purpureogenus* for producing gluconic acid were however less effective in bioleaching of REE than the bacterium *Gluconobacter oxydans*. Abiotic control experiments showed that an increasing concentration of gluconic acid from 10 to 90 mM could increase the total leaching efficiency of REE from 24% to 45%. The supernatants of *Gluconobacter* B58, containing an average 12.5 mM gluconic acid were more effective for solubilizing the REE from a fluid catalytic cracking (FCC) catalyst than abiotically prepared solutions of 30 and 90 mM gluconic acid (Figure 5) [36].

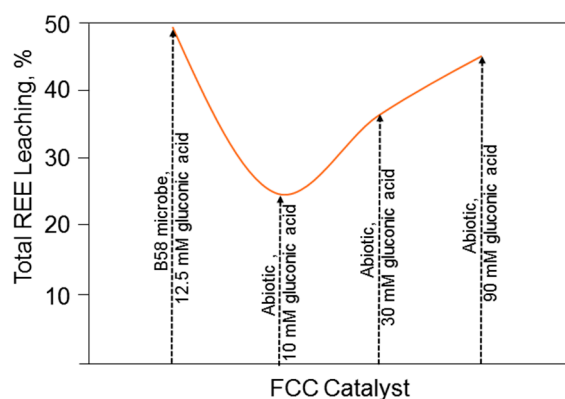


Figure 5. Leaching efficiency of total rare earth elements (REE) from fluid catalytic cracking (FCC) catalyst after 24 h using a four-day B58 culture supernatant producing 12.5 mM gluconic acid and abiotic solution of 10, 30 and 90 mM gluconic acid. Adapted from [36]. Copyright ELSEVIER, 2016.

The interactions of REEs with biogenic phosphates and specific transporter uptake are two recently disclosed processes having the potential to recover the REEs from aqueous waste streams. Firstly, the sorption on, or the encapsulation into the biogenic phosphate nano-minerals and the formation of REE-phosphate complexes were demonstrated using biofilms of *Serratia* strains [37,38]. Using an extremophile, acidophilic methanotrophic *Methylococcus* *fumarolicum*, Pol et al. [39] observed

that the function of methanol dehydrogenase (MDH) in this bacterium strictly depends on the presence of REEs. The analysis of protein structure and alignment of amino acid sequences has revealed that the REE-binding motifs in this particular MDH were also present in other MDH enzymes. This largely indicated that many MDHs are lanthanide dependent. It was assumed that by modifying the structure of protein, the specificity and binding efficiency of protein towards the substrate could be improved [40,41]. Henceforth, a REE-specific transporter to only recognize one particular REE could artificially evolve to concentrate REE in its cell. These approaches may be especially useful in recovering REEs from the aqueous waste streams.

Working with a mixed solution of La, Eu and Y, Texlier et al. [42] observed a preferential adsorption of europium ions onto the dead biomass of *Pseudomonas aeruginosa*. An investigation on the influential presence of metal cations viz. Al^{3+} , Ca^{2+} , Na^+ , K^+ and anionic salts viz. NO_3^- , SO_4^{2-} , Cl^- exhibited that the Al^{3+} was more inhibitive cation for adsorbing REEs by *Pseudomonas aeruginosa*. Nevertheless, the presence of thorium or uranyl ions could greatly decrease lanthanum adsorption (at $\text{pH} \approx 1.0$) by *Mycobacterium smegmatis* [43]. Fungal biomasses of *Rhizopusarrhizus*, *Saccharomyces cerevisiae*, *Aspergillus niger* were also examined for biosorption of REEs from various solutions. It has been observed that the presence of glutamate, sulfate, phosphate, chloride, carbonate and ethylenediamine tetraacetate (EDTA) in sorbate solution with metal ratios up to 15, can greatly affect the sorption of lanthanum onto fungal biomass [44]. Table 1 summarizes the R&D works for the biosorption of rare earth elements from various solutions.

Table 1. A summary of various biosorption processes of rare earth elements (T = temperature; t = contact time; B.D. = biomass dosage).

Microorganism	Composition of Treated Solution	Optimum Condition	Efficacy of Metal Recovery %	References
<i>Pseudomonas aeruginosa</i>	3 mM nitrate solution of: Ca, La, Eu, Yb and Gd	T , 20 °C; t , 2 h; pH, 5.0; B.D., 0.5 g/L	>90% Gd, and 77% Ca, La, Eu and Yb	[45,46]
<i>Pseudomonas putida</i>	10^{-8} M Ca, Sr, Eu, Zn, Cd, Hg in solution	T , 25 °C; t , 24 h; pH, 6.4; B.D., 0.075 mg/mL	90% Eu	[47]
<i>Bacillus subtilis</i>	Synthetic nitrate solution containing La	T , 25 °C; t , 30 min; pH, 4.0; B.D., 3 mg/mL	98% La	[48]
<i>Pseudomonas aeruginosa</i>	Chloride solution containing 0.25–2.0 mM REEs	T , 30 °C; t , 4 h; pH, 5.0; B.D., 0.124 g/L	Up to 97% REEs	[49]
<i>Shewanella putrefaciens</i>	Synthetic solution containing 0.14 mM Pm	T , 15 °C; t , 17 h; pH, 5.5; B.D., 5.6×10^{11} cells/L	90% Pm	[50]
<i>Mycobacterium smegmatis</i> and <i>Bacillus subtilis</i>	Synthetic Gd-nitrate solution	T , 20 °C; t , 1 h; pH, 5.0; B.D., 50 g/L	$\geq 98\%$ Gd	[45]
<i>Myxococcus xanthus</i> and <i>Saccharomyces cerevisiae</i>	0.4 mM nitrate solution of: La, Co, Mn, Pb, Ag, Zn, Cd	T , 28 °C; t , 1 h; pH, 4.5; B.D., 0.4 g/L solution	>95% La	[51]
<i>Mycobacterium smegmatis</i>	Nitrate solution of 30 mM Th, U, Yb, Eu, and La	T , 20 °C; t , 3 h; pH, 1.0–1.5; B.D., 50 g/L	99% REEs	[52]
(a) <i>Saccharomyces cerevisiae</i> ; (b) <i>Rhizopusarrhizus</i> ; (c) <i>Aspergillus terreus</i> ; (d) <i>Saccharomyces cerevisiae</i>	In synthetic solution (mg/L): 1.7 Sc, 7.9 Y, 2.0 Al, 1.7 Fe, 0.35 Ti, 2.1 Ca	T , 25 °C; t , 30 min; pH, 4.5; B.D., 0.5 g/L	(a) >98%; (b) 87%; (c) 87%; (d) 99% Y	[44]

Using bioaccumulation technique, an acidophilic fungal strain *Penidiella* sp., isolated through enrichment procedures to accumulate Dy, revealed the occurrence of REEs' accumulation onto the cell surfaces. Except for Sc, the strains were found to be accumulated with Dy, Nd, Y, Eu, Pr, Yb, and Tb. The fungal cells, however, did not accumulate the base metal ions viz. Mn^{2+} , Co^{2+} , Cu^{2+} , Al^{3+} , Zn^{3+} , Ga^{3+} [53]. The accumulation of lanthanum onto the surface of *Pseudomonas aeruginosa* inducing crystalline precipitation has been reported by Mullen et al. [48].

Regarding the bio-remediation of rare earths, Ce^{3+} and La^{3+} are primarily required in the oxidation of methanol in microbes thriving on plant leaves via utilizing the methanol exudates [54]. The presence of La^{3+} supports the enrichment and isolation of methylotrophic bacteria, and uses single carbon compounds for energy acquisition to growth. A better knowledge of the microbial role of REEs and their acquisition by microbes will certainly lead to the development of the bioremediation strategies for a clean-up job of rare earth's mine-sites and their recovery from divergent waste streams.

3.2. Bio-Reclamation of Precious Metals

Commercial bioprocessing units use oxidizing conditions and iron or sulfur oxidizers (such as *Acidithiobacillus ferrooxidans*, *Acidithiobacillus thiooxidans*, *sulfobacillus thermosulfidooxidans*) to generate ferric iron and sulfuric acid to accelerate metals dissolution via direct, indirect, thiosulfate and poly-sulfide mechanisms of biooxidation [24,55]. After initial removal of heavy metal impurities using biooxidation, the precious metals are subjected to biogenic cyanidation as depicted in Figure 6. Precious metals include palladium, Pd; platinum, Pt; rhodium, Rh; iridium, Ir; osmium, Os; ruthenium, Ru and gold, Au can be extracted through biogenic cyanide leaching of wastes/secondary materials. Glycine is a direct precursor of biogenic cyanide that can be produced via the oxidative decarboxylation of glycine performed by the cyanogenic bacteria. The biogenic cyanidation leaching has the advantages to replace the chemical cyanidation in gold leaching under an alkaline condition, making metal's recovery much more efficient, reducing the reagents and transportation costs, and minimizing the environmental issues of chemical cyanidation. This stems from the fact that the cyanogenic bacteria yielded cyanide from glycine for short periods at the late exponential or early stationary phase of its growth. The CN^- produced by *Chromobacterium violaceum*, a cyanogenic bacteria, is substantially used for gold recovery from the various waste streams, as with Equation (7) [26,55].

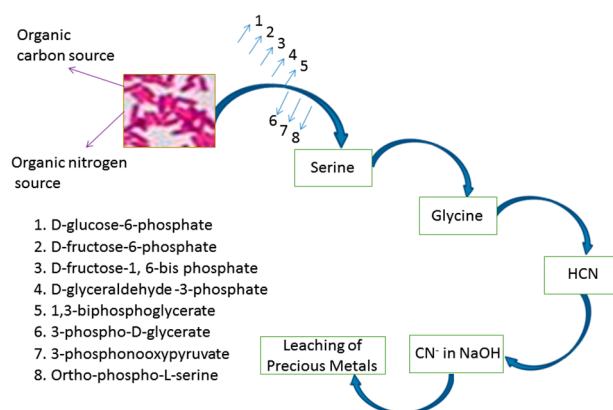
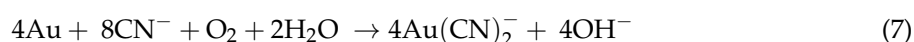


Figure 6. The production route of cyanide by microorganism with the subsequent use in cyanidation of precious metals (arrows are denoting the metabolic pathway of *Chromobacterium violaceum*).

In a preliminary study, Faramarzi et al. [26] reported the feasibility of gold recovery from waste printed circuit boards (PCBs) by *Chromobacterium violaceum*. The biogenic cyanidation of 5 mm × 10 mm sized PCBs, containing 10 mg of gold in each piece, could mobilize 15% gold as dicyanoaurate, $\text{Au}(\text{CN})_2^-$ in an alkaline solution. However, the microbe-mediated gold dissolution in presence of *Chromobacterium violaceum* with low metal mobilization did not show the formation of $\text{Au}(\text{CN})_2^-$ [56]. Further studies done by Brandl and Faramarzi [57] showed a better performance of *Chromobacterium violaceum* for mobilizing 68.5% gold in 7 days from waste PCBs. In contrast, *Pseudomonas fluorescens* and *Pseudomonas plecoglossicida* formed lower amount of dicyanoaurate. The gold complex was more

specifically unstable with *Pseudomonas fluorescens* due to the sorption of dicyanoaurate onto dead biomass or metal-cyanide's degradation [57]. A substantial amount of copper dissolution from e-waste was also obtained due to its high content in the waste and a faster kinetics of cyanidation. Recovery of other precious metals viz. silver and platinum from the e-waste was not very promising because of their resistance and toxic effects on microorganisms. Using wild culture of *Chromobacterium violaceum*, Chi et al. [58] obtained 13% gold and 37% copper in 8 days from mobile phone PCBs of content 0.025% Au and 34.5% Cu. They suggested an improved bio-reclamation of gold can be achieved by reducing the cyanide consumption from the copper containing material, and hence, proposed to substantially decrease copper content prior to undergo for biogenic cyanidation of precious metals. Recently, Tay et al. [59] could produce 70% more cyanide by metabolically-engineered strains of *Chromobacterium violaceum*, pTAC and pBAD in as compared to the cyanide generated by wild culture of *Chromobacterium violaceum*. Such a high amount of cyanide yielded more than twice the amount of gold leaching as well. Moreover, the cyanide production could be decoupled from quorum control by using the exogenous promoters. Modulations in proteome, resulting in a decreased cyanogenesis and an enhanced cyanolysis, may lead to a high amount of cyanide production. Moreover, the increase in cyanide production by the engineered strain does not change the protein levels of these enzymes, associated either directly or indirectly with the cyanolytic pathways. The study demonstrated that for the sustainable and effective bio-reclamation of precious metals from secondary sources, the genetically engineered microbes would be beneficial within their inherent capacity of cyanide production.

Other than the bacterial leaching of precious metals from secondary sources, bio-reduction of metal oxides to their metallic phase has also been studied in some cases. It has been observed that the microbial activity is helpful to generate an in situ reducing environment for metal ion reduction and the subsequent catalytic reactions as well [60–62]. Lenz et al. [63] showed the important role of metal species in reduction potential and bacterial uptake efficiency. For example, bio-reduction of various palladium species has shown the influence of reduction potential for obtaining the reduced Pd⁰ nanoparticles using microbes (viz. *Geobacter*), biofilms, and granular sludge [60,64]. In a continuous recovery of palladium from granular sludge in an up-flow anaerobic bed reactor, Pat-Espadas et al. [61] showed the practical feasibility of the bio-recovery. Using *Serratia* sp. biofilms, developed on the polyurethane foam, palladium was recovered as biofilm-Pd⁰ nanocatalyst from industrial wastes [65]. The biorecovery of Pd(II) occurs in three concomitant steps: (i) biosorption with biomass via interactions with surface functional groups viz. carboxyl, phosphoryl, and amines; (ii) initial reduction to form Pd⁰ nucleation sites; and (iii) the growth of Pd⁰ on nuclei surfaces via the autocatalytic reduction of Pd(II) [66]. The results were in line with microbial activity of enzymes such as formate dehydrogenase and hydrogenase used for the bioreduction of Pd(II) [67–70]. Subsequent studies demonstrated that the metabolic activity of microorganisms is not essential to form the bio-Pd⁰ [60,66]. The applicability of metal-reducing bacteria like *Shewanella alga*, *Desulfovibrio vulgaris* in manufacturing the nanoparticles of gold and platinum has also been explored, but yet to be explored with waste streams [71]. The uses of native and palladized *Desulfovibrio desulfuricans* in recovery of precious metals have shown selectivity for Au over Pt and Pd from an e-waste leachate [72].

Employing the biosorption technique for ruthenium recovery from a metal-plating effluent and process streams of acetic acid production by *Rhodopseudomonas palustris* and *Corynebacterium glutamicum* strains has given inspiring results. The experimental results showed that an acid pre-treatment of *Rhodopseudomonas palustris* cells could improve Ru-loading from 86 to 145 mg/g dry cells along with good selectivity over the co-existing metals Ni and Zn in effluent [73]. Polyethyleneimine (PEI)-coated *Corynebacterium glutamicum* fibers yielded 110 mg/g Ru-loading from an acetic acid solution. This result was a 16.5-fold higher efficiency than the commercially used ion-exchange resin [73], and 6.9-fold higher with the surface modified biomass blended with chitosan [74]. The R&D works performed on reclamation via biosorption of precious metals from various secondary sources have been summarized in Tables 2–5. By analyzing the presented Tables, it can be stated that the use of specific microbes along

with their surface modification strategies may lead to the selective binding of these critical metals from their waste streams of complex nature.

Table 2. A summary for the biosorption studies of gold by dead biomass of algae (T = temperature; t = contact time; B.D. = biosorbent dosage).

Algae	Composition of Treated Solution	Optimum Process Conditions	Efficacy of Gold Recovery %	Remarks	References
<i>Chlorella vulgaris</i>	A solution containing 10^{-4} M of each Cr^{3+} , Cu^{2+} , Zn^{2+} , Au^{3+} , Hg^{2+}	T , 30 °C; t , 12 h; pH, 2; B.D., 5 mg/mL solution	90%	Lab scale study; except the AuCl_4^- and $\text{Cr}(\text{H}_2\text{O})_6^{3+}$ all other metals taken in oxidized form	[75]
<i>Sargassum natans</i>	8.5–1000 mg gold with other metal ions UO_2^{2+} , Pb^{2+} , Zn^{2+} , Ag^+ in solution	T , 30 °C; t , 2 h; pH 2.5	420 mg/g gold	Various cationic species of gold was used in lab scale study	[76]
<i>Sargassum fluitans</i>	2.2 mM gold in cyanide solution	T , 30 °C; t , 4 h; pH 2.0; B.D., 20 mg/150 mL solution	97%	Gold ions, Laboratory Scale Laboratory Scale	[77]
Alginate cross-linked with CaCl_2	Synthetic solution containing 25–500 ppm gold as HAuCl_4	T , 25 °C; t , 4 h; pH, 2.0; B.D., 0.075 g/75 mL solution	98%	Colloidal gold Laboratory Scale	[78]
Alginate cross-linked with CaCl_2	Synthetic solution containing 25–500 mg/L silver as AgNO_3	T , 25 °C; t , 4 h; pH, 2.0; B.D., 0.075 g/75 mL solution	85%	Colloidal Silver Laboratory Scale	[78]
De-alginate Seaweed Waste	Synthetic solution of gold	T , 25 °C; pH, 3.0; B.D., 0.5 g/mL solution	92%	Colloidal gold Laboratory Scale	[79]

Table 3. A summary of the biosorption studies of gold and silver by dead biomass of fungi and yeast (T = temperature, t = contact time, B.D. = biosorbent dosage).

Microbe Used	Composition of Treated Solution	Optimum Process Conditions	Efficacy of Metal Recovery %	Remarks	References
<i>Cladosporium cladosporioides</i>	Synthetic solution of gold	T , 30 °C; t , 30 min; pH, 4.0; B.D., 0.05 g/25 mL	96.6 mg Au/g biomass dosage	Lab study recovering the colloidal gold	[80]
<i>Cladosporium cladosporioides</i>	Synthetic solution containing Ag^+	T , 30 °C; t , 30 min; pH, 4.0; B.D., 0.05 g/25 mL	15.2 mg Ag/g biomass dosage	-	[80]
<i>Rhizopus arrhizus</i>	8.5–1000 mg gold with other metal ions UO_2^{2+} , Pb^{2+} , Zn^{2+} , Ag^+ in solution	T , 30 °C; t , 30 min; pH 2.5; B.D., 0.5 mg/mL	92–95% gold	Lab study recovering the colloidal gold	[76]
PVA-immobilized biomass (<i>Fomitopsis carnea</i>)	10–100 mg/L gold solution	T , 25 °C; t , 33.3 h; pH, 1.0–13.0; B.D., 5 g/100 mL	~80% gold	Lab study recovering the colloidal gold	[81]
<i>Aspergillus niger</i>	2.5–1000 mg/L silver as AgNO_3	T , 30 °C, t , 1 h; pH, 7; B.D., 2 g/25 mL solution	75% silver	Lab study recovering the colloidal silver	[82]
<i>Neurospora crassa</i>	2.5–1000 mg/L silver as AgNO_3	T , 30 °C; t , 1 h; pH, 7.0; B.D., 2 g/25 mL solution	68.2% silver	Lab scale study recovering the colloidal silver	[82]
<i>Fusarium oxysporium</i>	2.5–1000 mg/L silver as AgNO_3	T , 30 °C; t , 1 h; pH, 7.0; B.D., 2 g/25 mL solution	57.5% silver	Lab scale study recovering the colloidal silver	[82]

Table 4. A summary of the biosorption studies of precious metals using protein biomass (T = temperature, t = contact time, B.D. = biosorbent dosage).

Protein Used	Composition of Treated Solution	Optimum Condition	Efficacy of Metal Recovery %	References
Hen egg shell membrane	1 g/L KAu(CN) ₂ , HAuCl ₄	T , 25 °C; t , 2 h; pH, 3.0; B.D., 0.25–0.35 g/50 mL	147 mg Au ⁺ /g 618 mgAu ³⁺ /g	[83]
Lysozyme from hen egg white	Cu-refining solution containing 82 g/L Au, 3.8 b/L Pt and 21 g/L Pd	T , 25 °C; t , 1 h; pH, 4.0; B.D., 2 mg/10 mL	165 g/kg Au; 11 g/kg Pt; 1 g/kg Pd	[84]
Alfalfa	0.3 mM of each metal ions: Au ³⁺ , Cd ²⁺ , Cu ²⁺ , Cr ³⁺ , Pb ²⁺ , Ni ²⁺ , Zn ²⁺	T , 30 °C; t , 1 h; pH, 5.0; B.D., 5 mg/mL	58% Au	[85]
Condensed-tannin gel	Synthetic solution of Pd	T , 25 °C; t , 3 days; pH, 2.0; B.D., 3–35 mg/150 mL	95% Pd	[86]
Bayberry tannin immobilized collagen fiber membrane	Chloride solution containing 48.8 mg/L Pd and Pt each	T , 40 °C; t , 4 h; pH, 4.0; B.D., 0.1 g membrane	97.4% Pd and 94% Pt	[87]
Acid-washed <i>Ucides cordatus</i> (waste crab shells)	Synthetic cyanide solution containing 2.2 mM gold	T , 25 °C; t , 24 h; pH, 3.4; B.D., 40 mg/20 mL solution	92% Au	[88]
Sulfur derivative of chitosan	Synthetic gold solution in 1 M HCl	T , 25 °C; t , 5 days; pH, 2.0–3.0; B.D., 8 mg/150 mL solution	400 mg Au/g B.D.	[89]
Glutaraldehyde crosslinked derivatives of chitosan (GCC, RADC)	Synthetic gold solution in 1 M HCl	T , 25 °C; t , 5 days; pH, 2.0–3.0; B.D., 8 mg/150 mL solution	600 mg Au/g B.D.	[89]
Thiourea derivative of chitosan	20 mg/L Pt solution	T , 25 °C; t , 72 h; pH, 2.0; B.D., 10 mg/100 mL solution	120 mg Pt/g B.D.	[90]
Glutaraldehyde crosslinked chitosan	50 mg/L Pd and 20 mg/L Pt	T , ~20 °C; t , 72 h; pH, 2.0; B.D., 10 mg/100 mL solution	400 mg Pd and 120 mg Pt/g B.D.	[91,92]
Chitosan derivatives	10–20 mg/L Pd and 20–40 mg/L Pt	T , ~20 °C; t , 3 days; pH, 2.0; B.D., 3–35 mg/150 mL solution	1.3 mM Pd and 0.8 mM Pt/g B.D.	[93]

Table 5. A summary of the biosorption studies of precious metals using dead bacterial biomass (T = temperature, t = contact time, B.D. = biosorbent dosage).

Bacteria Used	Composition of Treated Solution	*Optimum Condition	Efficacy of Metal Recovery %	References
<i>Streptomyces erythraeus</i> and <i>Spirulina platensis</i>	1 g/L KAu(CN) ₂ , HAuCl ₄ H ₂ O	T , 25 °C; t , 2 h; pH, 3.0; B.D., 0.25–0.35 g/50 mL	99% and 98% Au, respectively	[83]
<i>Desulfovibrio desulfuricans</i>	Synthetic solution: 5–50 mg/L Pd and Pt	T , 30 °C; t , 4 days; pH, 3.0; B.D., 50 mg/L solution	90% Pd and 92% Pt	[93]
<i>Bacillus subtilis</i>	Synthetic CN [−] containing 2.2 mMAu	T , 30 °C; t , 4 h; pH, 2.0; B.D., 40 mg/20 mL solution	97% Au	[77]

3.3. Bio-Reclamation of Other Critical Metals

Due to the extensive uses of selenium and tellurium in electronics, semi-conductors and solar cells, their scarcity and uneven distribution, it necessitates the recovery methods of these metals from waste materials/solutions [94,95]. The microorganism *Pseudomonas mendocina* can reduce tellurium from its higher oxidation states (+2, +4 and +6) to Te⁰ [96,97]. In an integrated process of recovery of critical metals from used CdTe solar cells, Rajwade and Paknikar [96] firstly employed fungal strains for selective removal of silver and cadmium from acidic solution. Then, *Pseudomonas mendocina* could lead to reductive precipitation of Te⁰. In a recent study, Te⁰ was directly recovered using *Pseudomonas* sp. (strain EPR3) from the divergent sources, anode slime of copper, and TeO₂/CdTe/Bi₂Te₃ used

material [97]. The occurrence of several inter-related and coupled processes was observed during the bacterial speciation of tellurium. In the presence of microbial cultures, the soluble metal species diffuse across the cell membrane, inside which a reversible redox reaction simultaneously takes place with the methylation. As a result of this, an internal bacterial precipitation of Te^0 occurs. Some of the gaseous species of tellurium diffused out through the cell membrane which may volatilize to atmosphere or reform the soluble tellurite by following the Challenger mechanism [97]. However, the microbial volatilization of metals like Te and Se poses serious risks to the environment and human health.

Uranium from its resources including the waste materials is processed by chemical/bacterial leaching in sulfuric acid by following Equation (8) below.



Extraction of uranium from various sources has been a long-term quest due to its strategic importance in defence, and civil nuclear industry. The anthropogenic activities from mining to generation of spent fuel have led to uranium contamination in the environment [98–100]. Therefore, the microbial strategy has been explored not only to reclaim but also for remediation purposes. The interactions between uranium and bioorganism involve different mechanisms, certainly not limited to biosorption on the cell surface, bioprecipitation as hydrogen uranyl phosphate (HUO_2PO_4), bioaccumulation in polyphosphate granules, and bioreduction as uraninite (UO_2). Bio precipitation of uranium as HUO_2PO_4 from waste streams using bacterial phosphate has long been studied for hydrolytic release of inorganic phosphates [100]. To improve the economy of bioprecipitation, Paterson-Beedle et al. [38] used phytic acid, a waste product of plants as the phosphate source in the system. Furthermore, the phosphatase activity of microbial cultures was improved by the overexpression of phosphatase through genetic engineering. The bacterial reduction of uranium from the +6 to +4 state is a potent method of bioremediation for in situ immobilization and remediation of subsurface environment [101]. This approach is cost-effective compared to the pump and treat remediation; however, supply of electron donors, maintenance of proper redox environment and long durability of bio reduced uranium (+4) precipitates have been found to be challenging. The exploitation of 349 mg U/kg black shale using *Acidithiobacillus ferrooxidans* has been investigated by comparing the effect of inoculation and non-inoculation reactor [102]. It was observed that the batch-type reactor inoculated with microbial cells exhibited a higher redox potential, lower pH and higher quantity of ferric ions than that which was observed for the non-inoculated reactor until 200 h. The condition facilitated by microbial activity could efficiently improve the bioleaching of uranium; however, with sufficient nutrients and Fe^{3+} as energy source, the indigenous microorganisms in non-inoculated black shell could also be activated over time. The study revealed a scope of cost reduction by the means of lower consumption of Fe^{2+} and nutrients in the system [102].

Rhenium, one of the energy-critical elements has a prime use in Pt-Re catalysts [103]. A study on rhenium extraction from spent Pt-Re catalyst using *Bacillus megaterium* as a cyanogenic bacterium has been reported by Motaghd et al. [104]. The response surface methodology was applied by keeping pulp density and initial glycine concentration as two main influential parameters for extracting the critical metals; however, it was found to be better only to extract the Re. After seven days of bioleaching under the optimal condition of 40 g/L pulp density and 12.8 g/L initial concentration of glycine, 98% Re could be leached with only 16% Pt in leach liquor. During the bioleaching, an increase in free cyanide concentration to 3.6 mg/L under the varied pH (6.7 to 9.0) and dissolved oxygen concentration (2 to 5 mg/L) clearly demonstrated the growth characteristics of *Bacillus megaterium*. A lower leachability of Pt can be attributed to its resistance to being mobilized using the cyanogenic microorganisms [26,105]. Recovery of Re from a waste water employing biosorption technique has also been studied. Using the brown algae *Laminaria japonica* in its modified cross-link gel structure exhibited a maximum 97% sorption with the capacity 37.2 mg/g at pH 6.0 [106]. The result showed a higher affinity for Re in comparison with the earlier used sorbents of persimmon and orange peel [107,108]. This indicated the positive effect of H_2SO_4 -modified structure on rhenium adsorption that was resultant of the increased surface area of the sorption sites on algae cell walls.

4. Prospects for the Bio-Reclamation of Critical Metals from Secondary Resources

The stringent regulations on environmental pollutions and constraints on practicing the current operations will create demand for new methodologies to recover the critical metals from waste streams, providing an ideal situation for bio-reclamation approaches. Notably, the particular microbe-to-REE interactions and leaching mechanism have hardly been focused upon and we are facing challenges as their bio-geo-chemistry is stable towards the redox transformations. For such a challenging task, a multidisciplinary approach of various research communities belonging microbiology, metallurgy, and chemistry to unfold the complexation of critical metals with biogenic lixivants and/or their enrichment/separation using the bio-accumulation/sorption/reduction would be much needed.

Reclamation of critical metals by heterotrophic bioleaching is far less discussed, despite its distinctive potential. Heterotrophic bio-reclamation using indigenous fungal species could be feasible to recover REEs from solid wastes. Biogenic organic acids coupled with biogenic chelators could be efficiently employed to liberate the REEs from divergent wastes. Extensive uses of microorganisms in leaching of precious metals have been restricted by the limited exploitation of cyanogenic microbes only to *Chromobacterium violaceum*. Therefore, the construction of a metabolic engineered strain of related microbial cultures can produce more cyanide in lixiviant that may enhance the recovery by several fold compared to using the wild culture. Immobilization of bacterial cells on supports, like polyurethane foam and design of integrated-column operations can lead to efficient recovery of metals. The critical metals able to form insoluble phosphates can be amenable for recovery via the precipitation-crystallization route through the biogenic production of high phosphate ligands locally. Employing the biosorption technique, the microbial biomass and its constituents (viz. bacteria, yeast, fungi, algae, derived from activated sludge or fermentation waste, including fast growing microorganism crab shells and seaweeds) can be potential sorbent material for critical metals recovery.

Overall, the most important factor is the practical implementation of lab scale technology to the actual systems. Indeed, microbe-to-metal interactions have been efficiently employed in exploitation of primary ores, but are not sufficiently investigated to be commissioned for reclamation of strategic and energy critical metals from secondary sources. In an actual system, the conditions of waste streams like extreme acidic/alkaline pH, high amount of total dissolved solids, strong ionic strength, presence of organics and undesired toxic metals are the bottlenecks which can deviate the reclamation efficiency of critical metals, because most of the lab studies are performed under ideal conditions using synthetic solutions. Notably, critical/scarce metals in the waste solutions remain in trace amounts, sometimes as micro-sized colloids, and are sorbed onto suspended materials or as complex compound in solid wastes like alloys, scraps, metallurgical slags; hence, bio-reclamation should furthermore focus on “unfavorable” conditions and combination of metals that cannot be separated well by either the pyro or hydro(chemical) route of metallurgy.

5. Conclusions

Bio-reclamation of critical metals from the secondary/urban mine resources can be attractive in economic and technical terms. The microbial approach does not require high capital to invest, energy for smelting or electro-reduction; and a minimized generation of new waste compared to the conventional extraction route of REEs and other critical metals are major advantages which can mitigate their supply risk in a sustainable manner. However, the complex nature of secondary materials with a minor amount of critical metals compared to the other co-engineered metals is the most difficult challenge. Bioleaching of REEs from the urban mined sources is quite different than using the phosphate solubilizing bacteria in the case of primary ores, despite the organic acid producing microorganisms being useful to leach the REEs. In the case of precious metals like Au, Ag, Pt, and Pd, the biogenic cyanidation is the promising one. Reclaiming the precious metals of very low concentration from various waste stream solutions, biosorption and bioaccumulation techniques have been employed frequently to get the prominent recovery of critical metals. In some cases, like for Pd and Te, the bio-reduction has also been reported to directly recover the critical metals in nano-sized

metallic form. Nevertheless, the reported literature on bio-reclamation critical metals belongs to only lab scale basic research; therefore, a lot of efforts are needed to get the real benefit of microbial strategy for treating the urban mined secondary materials.

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