

Uranium bioleaching from low-grade carbonaceous-siliceousargillaceous type uranium ore using an indigenous *Acidithiobacillus ferrooxidans*

Xuegang Wang^{1,2} · Peng Li² · Yajie Liu² · Zhanxue Sun² · Liyuan Chai^{1,3} · Xiaobo Min^{1,3} · Yadan Guo² · Zhihong Zheng² · Yong Ke^{1,3} · Yanjie Liang^{1,3}

Received: 12 April 2018/Published online: 19 June 2018 © Akadémiai Kiadó, Budapest, Hungary 2018

Abstract

We evaluated the effectiveness of bioleaching uranium from a low-grade carbonaceous-siliceous-argillaceous type uranium ore using an indigenous iron-oxidizing bacteria, *Acidithiobacillus ferrooxidans*, isolated from local uranium ore. The effects of initial acidity, pulp density and ferrous ion concentration of the feed solution were investigated. The uranium (U_3O_8) content was 0.036% by weight. Using uranium ore acidified leachate as medium with initial ferrous ion concentrations of 3 g/L, pH 1.7 and pulp density of 20% as optimal conditions, the maximum rate of dissolved uranium recovery was 85.14%. This approach is thus, suitable for recovering uranium from low-grade CSA ore using bacterial leaching.

Keywords Uranium bioleaching · Low-grade · Carbonaceous-Siliceous-Argillaceous uranium ore · *Acidithiobacillus ferrooxidans*

Introduction

Carbonaceous-siliceous-argillaceous (CSA) type uranium deposit is one of the four types of uranium deposits in China and is an unconventional uranium resource. It has the properties of shale stability, large thickness and complex composition, with an abundance of organics, pyrite, clay, silica and phosphorus, a typical uranium content of 0.03–0.05% by weight [1, 2]. For CSA uranium ore leaching, the chemical acid extraction method is widely used in large operations. However, this process is accompanied with significant disadvantages, such as high

Xiaobo Min mxbcsu@163.com

- ¹ Institute of Environmental Science and Engineering, School of Metallurgy and Environment, Central South University, Changsha 410083, Hunan, China
- ² State Key Laboratory Breeding Base of Nuclear Resources and Environment, East China University of Technology, Nanchang 330013, Jiangxi, China
- ³ Chinese National Engineering Research Center for Control and Treatment of Heavy Metal Pollution, Changsha 410083, Hunan, China

costs due to the addition of sulfuric acid and an oxidant (usually MnO_2 , H_2O_2 , or $KClO_3$), environmental noncompatibility and potential ecological risks [3–7]. Therefore, the development of a new strategy that focuses on an effective and environmentally-sound leaching process of low-grade CSA uranium ore is necessary.

The process of bioleaching has the potential to save costs, which is environment friendly and can recycle resources for the hydrometallurgy of metal ores [8, 9]. The mechanism of the uranium bioleaching process is that the bacteria oxidized pyrite (FeS₂) and soluble Fe²⁺ to Fe³⁺ as the leaching oxidants detailed in Eqs. (1–2), Fe³⁺ readily attacks minerals incorporating U⁴⁺ to oxidize it to U⁶⁺ according to Eq. (3).

 $FeS_2 + 7O_2 + 2H_2O \rightarrow 2FeSO_4 + 2H_2SO_4 \tag{1}$

$$4\text{FeSO}_4 + \text{O}_2 + 2\text{H}_2\text{SO}_4 \rightarrow 2\text{Fe}_2(\text{SO}_4)_3 + 2\text{H}_2\text{O}$$
(2)

$$UO_2 + Fe_2(SO_4)_3 \rightarrow UO_2SO_4 + 2FeSO_4$$
 (3)

The Fe²⁺ can be re-oxidised by microbes to Fe³⁺, which takes part in the oxidation process again. In an acidic solution Fe²⁺ is stable and the leaching mediated by Fe³⁺ would be slow. The bio-oxidation is about $10^{5}-10^{6}$ times faster than the chemical oxidation thus enhancing the

uranium dissolution rate. It is believed to be one of the most promising technologies because it consumes less acid, does not have a requirement of an oxidant additive, is a simple process with no need of sophisticated operator training and it can successfully treat certain mineral assemblages not amenable to treatment by conventional processes [10]. Moon-Sung Choi [11] reported the feasibility of bacterial recovery of uranium from low-grade black schists from the Okcheon District, Korea, using A. ferrooxidans. The uranium recovery reached 80%, while only 18% was leached without involving microbial activity. Over the past few decades, the bioleaching process has been applied for the extraction of copper [12-14], zinc [15, 16], gold [17], uranium [18–21], multiple metals [22, 23] and precious metals from printed circuit boards [24–26]. Therefore, this technique is proving to be a profitable alternative to the conventional processes.

With the rapid development and construction of China's nuclear power program, the demand for uranium resources has increased dramatically. Based on the view that CSA uranium ore and other unconventional uranium resources would make a positive contribution to the sustainable supply of uranium resources, officials have already begun to pay attention to the exploitation of uranium from CSA uranium deposits in China and many other countries and regions. In this study, the feasibility of uranium bioleaching from low-grade CSA uranium ore was explored. The effects of critical parameters was comprehensively studied in batch experiments.

Experimental

Preparation of uranium ore samples

The CSA uranium ore used in this investigation was collected from a local uranium mine in the Xiushui District, Jiangxi Province, China. For experimental use, the ore samples were prepared in a three-stage crushing operation. Initially, the uranium ore samples were manually cut into small pieces (diameter < 50 mm) using stainless steel blades. Next, the ores were crushed by a vibration jaw crusher. Finally, the uranium ores were sieved using a vibrator shifter through 16 mesh screen (diameter < 1 mm) dried to a constant weight in a drying oven at 105 °C and used for all the leaching experiments.

Microorganism isolation, culturing and domestication

The bacteria strain used in the experiments was *A. ferrooxidans*, which was isolated from the CSA uranium ore. The isolation was effected by using a dilution/enrichment technique in ferrous sulphate medium followed by singlecolony isolation on silica gel solid medium. 16S rDNA sequencing and phylogenetic analysis [27] identified the isolate as an indigenous acidophilic strain *A. ferrooxidans*, which belongs to the autotrophic bacteria. Specifically, it was an aerobic microorganism capable of oxidizing Fe^{2+} and S to obtain the energy required for growth.

The bacteria strain of A. ferrooxidans was cultured and domesticated in a modified 9 K medium composed of CSA uranium ore. The composition of the modified 9 K medium was as follows: 24.8 g of FeSO₄·7H₂O as an energy source, 3.0 g of (NH₄)₂SO₄, 0.5 g of K₂HPO₄, 0.1 g of KCl, 0.01 g of Ca(NO₃)₂·4H₂O, 0.5 g of MgSO₄·7H₂O, 0.02% (w/v) yeast extract and 1000 mL distilled water. A. ferrooxidans was inoculated at 20% (v/v) in a 250 mL Erlenmeyer flask containing 100 mL of 9 K medium, 20 g CSA uranium ore, an initial Fe²⁺ concentration of 5.0 g/L and the initial pH adjusted to 1.8 using 5.0 M sulfuric acid. The culture was incubated at 30 ± 1 °C on a constant-temperature rotatory shaker at 120 rpm. Bacteria inoculation would be repeated until the oxidation rate of Fe^{2+} reaches 90%. The bacteria stocks used in the leaching experiments were adapted to the CSA uranium ore through continuous subculturing in the way described above until a constant iron oxidation rate was achieved.

Bioleaching experiments

In order to investigate the influence of sulfuric acid content, pulp density and initial Fe²⁺ concentration, a series of flasks were arranged with sulfuric acid content at 1 g/L, 3 g/L, 5 g/L, 10 g/L, 20 g/L and 30 g/L; pulp density at 10% (w/v), 20%, 30%, 40% and 50%; initial Fe²⁺ concentrations of 0 g/L, 1 g/L, 3 g/L and 5 g/L. The bioleaching experiments were conducted using a 250 mL triangular flask and incubated at 30 °C with shaking at 120 rpm. At different elapsed times, samples were collected and filtered through 0.22 µm membranes. The samples were analyzed for pH, redox potential (Eh), uranyl-VI (U⁶⁺), ferric ion (Fe³⁺) and ferrous (Fe²⁺) concentration.

Analytical methods

The chemical composition of uranium ore sample was analyzed using a wavelength-dispersive X-ray fluorescence spectrometer (WD-XRF, Tiger S8, Germany) described in our previous study [28]. The pH and Eh (redox potential) were measured using a pH meter (ST3100, Ohaus, USA). The concentration of ferrous and total iron concentration in solution were determined by titration with EDTA. Concentration of the ferric iron was calculated by subtracting the concentration of ferrous from the total iron. Uranyl ions in solution were assayed using titanium reduction-ammonium vanadate titration [18]. The surface characteristics and morphology of the bioleaching residues were characterized by scanning electron microscopy (SEM, QuantaTM 250, FEI, USA). The X-ray diffraction (XRD, Bruker D8, Germany) characterizations were conducted for the phase composition detection.

Results and discussion

The mineral components of uranium ore

The chemical composition of raw uranium ore sample is listed in Table 1.

Analysis of the sample using XRD indicated that quartz (SiO_2) , pyrite (FeS_2) and hematite (Fe_2O_3) were the dominant component present and that it contained small amounts of other impurities including illite $(KAl_2[(Al,Si)Si_3O_{10}] (OH)_2 \cdot nH_2O)$, plagioclase $(Na(AlSi_3O_8), Ca(Al_2Si_2O_8))$ and kaolinite $(Al_4(Si_4O_{10})(OH)_8)$.

The growth curve of A. ferrooxidans

Analysis of the ferrous oxidation rate of *A. ferrooxidans* in modified 9 K medium was conducted under conditions of pH 1.8, an initial Fe^{2+} concentration of 5 g/L and a pulp density of 20%. The growth curve of *A. ferrooxidans* is shown in Fig. 1.

As it depicted in Fig. 1, the ferrous oxidation rate of the bacteria was very slow from 0 to 50 h, bacteria grew in the lag phase and adapted to the new leaching environmental conditions. The bacteria steadily grew from 50 to 105 h and the oxidation rate of Fe^{2+} reached approximately 2.5 g/(L·d) in 105 h. Bacteria with high oxidation activity

 Table 1 Chemical composition

 of raw low-grade CSA uranium

 ore sample

Component	Contents (wt. %)
SiO ₂	60.54
Al_2O_3	7.16
Fe ₂ O ₃	4.00
FeO	0.236
CaO	0.123
MgO	5.02
K ₂ O	1.77
Na ₂ O	0.037
TiO ₂	0.504
MnO	0.066
P_2O_5	0.2
CO_2	0.18
Corganic	13.01
U	0.036





6

Fig. 1 Ferrous oxidation rate of bacteria in modified 9 K medium with time (conditions: 1.8 pH, 5 g/L Fe²⁺, 20% PD, 30 °C, 120 rpm)

were obtained after 378 h of domestication and culturing. The oxidation rate of Fe²⁺ was greater than 4.3 g/(L·d), which indicated that *A. ferrooxidans* adapted remarkably well to the CSA uranium ore leaching solution following its acclimatizing culture. Good viability and ferrous oxidation capacity was gradually discovered during the subsequent bioleaching process.

Effect of sulfuric acid concentration

Acidified sulfuric acid concentration (or pH) is a crucial parameter for bacterial leaching of metals. In this study, the additive content of sulfuric acid was controlled at 1 g/L, 3 g/L, 5 g/L, 10 g/L, 20 g/L and 30 g/L. The pulp density was 40% (w/v) and the initial Fe²⁺ concentration was 5 g/L. The uranium leaching rate and pH change with respect to different initial sulfuric acid concentrations are depicted in Fig. 2.



Fig. 2 Uranium leaching rate and pH with respect to different initial sulfuric acid concentration (other conditions: 5 g/L Fe²⁺, 40% PD, 30 °C, 120 rpm)

As shown in Fig. 2, an increase in sulfuric acid content lowered the pH of the system. An opposite trend for the uranium leaching rate vs. the initial acid content was observed. When the acid content ranged from 3 g/L to 5 g/L, the uranium leaching rate improved significantly and the corresponding pH was 1.77 to 1.54. Under conditions of relatively low pH (< 2.0), the formation of jarosite through iron precipitation is inhibited [27]. The solid state of the uranium component should be efficiently oxidized to the dissolved form UO_2^{2+} by the ferric iron formed during microbial metabolism. As the acid content increased from 5 g/L to 30 g/L, the uranium leaching rate slowed. Any positive effect of acid consumption on the uranium leaching rate was negligible. The pH decreased from 1.54 to 0.76. The results indicated that higher concentrations of sulfuric acid could inhibit microbial growth and hinder the oxidation of U^{4+} to U^{6+} , which resulting in the increase of uranium leaching rate is slow and other problems associated with the dissolution of impurities. Furthermore, the increase in acid consumption would increase the investment costs.

A suitable acidity would reduce the extra mineral dissolution from the surrounding rocks, thus decrease the content of sulfuric acid and maintained the pH conditions for the microorganism metabolism. It has been documented [20] that *A. ferrooxidans* displays the highest uranium extraction capacity when the pH is less than 2.0. On the contrary, when the pH is greater than 2.0, a side reaction of the formation of jarosite occurs. This results in the ore surface being covered by the generated jarosite particulates and the uranium extraction is inhibited, leading to a decrease in the uranium extraction efficiency of the entire system.

Effect of pulp density

Experiments at different pulp densities of 10% (w/v), 20%, 30%, 40% and 50% were conducted under operating conditions of pH 1.7 and an initial Fe²⁺ concentration of 5 g/L (Fig. 3). The inoculated bacteria were in a 100 mL solution with a Fe²⁺ oxidation efficiency of 95%.

As shown in Fig. 3, Fe^{3+} and U^{6+} increased in relation with the increase in pulp density, while the uranium leaching rate underwent an opposite response. The reason for this phenomenon may be at low pulp densities, uranium minerals were easily extracted to the aqueous solution, which contributed to the high uranium leaching rate. However, because of the limitation of total uranium content in the added ore, U^{6+} concentration remained at a low level. The Eh value of the leaching solution was greater than 400 mV at all the pulp densities (Fig. 3b), which indicated that pulp density had no effect on the Eh of the leaching solution. A lower pulp density would be favorable for a high uranium leaching efficiency. However, the amount of supplied water required would be increased. Also, along with the disadvantage of the lack of bacterial concentration and the intense acid consumption, the low pulp density also led to a low uranium concentration in the leaching liquid, which increased the difficulty of concentrating the leached uranium. Therefore, the dominant pulp density was determined accoring to a comprehensive consideration of the uranium leaching rate, the consumption needs of water and the amount of inoculated bacterial.

As shown in Fig. 3c, when the pulp density was 10% and 20%, the concentration of Fe^{3+} 5 g/L, the leaching efficiency was 88.33% and 84.17%, respectively. The uranium concentration in the leaching liquid was about 31 mg/L and 60 mg/L, respectively. This indicated that a pulp density ranging from 10% to 20% was most beneficial for uranium bioleaching. When the pulp density was greater than 30%, the concentration of Fe^{3+} in the system reached 7.17 g/L and the uranium leaching efficiency decreased to 70% (Fig. 4c). The high pulp density had a negative effect on the uranium bioleaching process. This may be ascribed to the improved shear force generated from the high pulp density, which was detrimental to the immobilization of the bacteria on the uranium ore surface. On the other hand, at a high pulp density the dissolved Fe³⁺ concentration increased dramatically and the components that increased the utilization of acid increased, the pH to easily exceed 2.0. This situation caused the precipitation of Fe³⁺ on the uranium ore surface, which further inhibited the dissolving of the uranium. On the other hand, A. ferrooxidans is noted to be a microorganism that uses carbon dioxide as carbon source and oxygen as an electron acceptor; so, under the high pulp density conditions, aggregated fine ore particles would be formed and the mass transportation of carbon dioxide and oxygen would be inhibited [11, 29]. Therefore, in subsequent experiments, a 20% pulp density was maintained.

Effect of ferrous ion concentration

Ferrous ion was the energy source for the uranium bioleaching process. The suitable Fe^{2+} concentration favored the growth of the microorganism and ensured an improved microorganism concentration and oxidation capacity. The cultured *A. ferrooxidans* was put in three triangle bottles each containing 20 g of uranium ore. The pulp density was 20% and the pH was 1.7. To investigate the influence of ferrous ion concentration on the uranium bioleaching efficiency, the initial ferrous ion concentration was controlled at 0 g/L, 1.0 g/L, 3.0 g/L and 5.0 g/L.

As shown in Fig. 4a, the oxidation rate of ferrous ion was increased with the higher concentrations of the ferrous ion. After leaching for 245 h, the ferrous ion concentration



Fig. 3 Ferric iron concentration (a), redox potential (b), uranium leaching rate (c) and uranium concentration (d) under different pulp density with respect to time (other conditions: 1.7 pH, 5 g/L Fe²⁺, 30 °C, 120 rpm)

decreased to 0.12 mg/L due to its consumption by the microorganisms (Fig. 4a). When the initial ferrous ion concentration was 0 g/L, the Eh was 460 mV (Fig. 4c) and the corresponding uranium bioleaching efficiency was only 68.89% (Fig. 4d). This indicated that a low ferrous ion concentration would result in a lower oxidation capacity and therefore to a lower uranium bioleaching efficiency.

The concentration of ferric iron and the Eh values were increased with the higher initial ferrous ion concentrations (Fig. 4b, c). When the ferrous ion concentration was 3 g/L and 5 g/L, the Eh value was maintained at 500 mV, the ferric iron concentration was 6.68 g/L and 8.20 g/L, respectively. This indicated that the ferric iron concentration had less of an effect on the Eh value and the leaching efficiency when the ferric iron concentration was greater than 6 g/L.

The highest uranium leaching efficiency was obtained when the initial ferrous ion concentration was 3 g/L (Fig. 4d). This may be ascribed to the strongest bacterial activity and the highest capacity for the biological metabolism of carbon dioxide at the 3 g/L Fe²⁺ concentration. When Fe²⁺ concentration was 5 g/L, Fe³⁺ concentration of leaching solution reached 8.2 g/L, a decrease in the uranium leaching efficiency was observed (Fig. 4d). This indicated that at high Fe²⁺ concentrations, Fe²⁺ tended to be oxidized to Fe³⁺ by the microorganisms and precipitated onto the ore surface [27, 29]. It is clear that the formed precipitates would inhibit further dissolving of the uranium ores. Similar studies [30, 31] have established that only small quantities of ferrous ion and ferric iron are needed to maximize uranium solubilization.

SEM and XRD analyses of the leached residues

Representative results from SEM images of the leached residues from the different bioleaching conditions are



Fig. 4 Ferrous iron concentration (a), ferric iron concentration (b), redox potential (c) and uranium leaching rate (d) under different initial concentration of Fe^{2+} with respect to time (other conditions: 1.7 pH, 20% PD, 30 °C, 120 rpm)

shown in Fig. 5. It can be seen that the surfaces of the leached residues were wrapped by a layer of reaction product, which inhibited the movement and diffusion of the mineral surface bacteria and leaching agents in mineral fissures [21, 32, 33]. The layer on the residue obtained from the bioleaching with 5.0 g/L of additional external Fe²⁺ ions was dense surface and less porous than that with 3.0 g/L of additional external Fe²⁺ ions. The addition of Fe²⁺ ions accelerated the uranium bioleaching, while the concentration of additional Fe²⁺ over 3.0 g/L weakened this positive effect and even hindered the dissolution of the uranium.

The XRD data of the leached residues after bioleaching, at the different leaching conditions is presented in Fig. 6. It can be seen that the XRD of the leached residues after 245 h show that jarosite ($KFe_3(SO_4)_2(OH)_6$) formed during the bioleaching process. The lower uranium leaching rate may have been related to the passivation layer formed on the surface of mineral particles, which may have hindered the uranium bioleaching. It can also be seen that with additional 5.0 g/L Fe²⁺ ions, the intensity of peaks corresponding to jarosite were stronger than that observed with additional 3.0 g/L Fe²⁺ ions. This suggested that there was more jarosite produced on the surface of particle, which then inhibited the dissolution of the uranium. The results of the mineralogical analysis by XRD were also confirmed above SEM observations.

Conclusion

The native culture of the mesophilic bacteria *A. ferrooxidans* isolated from a source of mine water showed good potential for the bioleaching of uranium from the low-



Fig. 5 SEM images of the bioleaching residues: (a) 3.0 g/L and (b) 5.0 g/L of ${\rm Fe}^{2+}$ ions

grade CAS uranium ore. The surrounding rock of the uranium ores used in this work was acidic, which contains less acid consuming gangue minerals. When the addition of sulfuric acid concentration was 5 g/L, the leaching solution pH was below 2.0, iron precipitation could be avoided. With the pulp density ranges from 10 to 20%, relatively high uranium bioleaching efficiency was obtained and the pulp density had almost no effect on bacterial growth or the final uranium leaching efficiency. Ferrous ion is the energy source for A. *ferrooxidans*. The additional external Fe^{2+} ions over 3 g/L resulted in the uranium leaching efficiency to decrease. SEM and XRD analyses of the leaching residues revealed that a product layer composed of jarosite was formed on the mineral surface when the concentration of additional Fe^{2+} was over 3 g/L, which resulted in a low uranium leaching rate during the later phase. A bio-recovery of 85.4% uranium in the leaching liquid was achieved in 10 days with the ore-adapted culture under the optimal conditions of pH 1.7, 20% pulp density, 30 °C, 120 rpm and with 3 g/L of additional Fe²⁺ ions. Furthermore, these experiments showed that the bioleaching



Fig. 6 XRD patterns of the bioleaching residues: (a) 3.0 g/L and (b) 5.0 g/L of Fe^{2+} ions

process can be more effective and economical for uranium extraction from low-grade CSA type uranium ores than other processes that are currently available.

Acknowledgements This work was financially supported by the National Nature Science Foundation of China (51564001, 41772266, 51764001), the Natural Science Foundation of Jiangxi Province (20161BAB213091, 20171BAB213019) and Project of State Key Laboratory Breeding Base of Nuclear Resources and Environment Fundamental Science (Z1602).

References

- Hu M, Yang S, Shao F (2012) Characteristics and resource potential of carbonaceous-siliceous-argillitic rock-hosted uranium depositis in xiushui area. J East China Inst Technol 35(2):129–135
- Zhao F (2012) An review on geology study of carbonaceoussiliceous-pelitic rock type uranium deposit in China and the strategy for its development. Uranium Geol 25(2):91–97

- Bhargava SK, Ram R, Pownceby M, Grocott S, Ring B, Tardio J, Jones L (2015) A review of acid leaching of uraninite. Hydrometallurgy 151:10–24
- 4. Gilligan R, Nikoloski AN (2015) The extraction of uranium from brannerite: a literature review. Miner Eng 71:34–48
- 5. Li YC, Min XB, Chai LY, Shi MQ, Tang CJ, Wang QW, Liang YJ, Lei J, Liyang WJ (2016) Co-treatment of gypsum sludge and Pb/Zn smelting slag for the solidification of sludge containing arsenic and heavy metals. J Environ Manag 181:756–761
- Sun Y, Wang X, Ai Y, Yu Z, Huang W, Chen C, Hayat T, Alsaedi A, Wang X (2017) Interaction of sulfonated graphene oxide with U(VI) studied by spectroscopic analysis and theoretical calculations. Chem Eng J 310:292–299
- Peng B, Tang X, Yu C, Xie S, Xiao M, Song Z, Tu X (2008) Heavy metal geochemistry of the acid mine drainage discharged from the Hejiacun uranium mine in central Hunan. China Environ Geol 57(2):421–434
- Anjum F, Shahid M, Akcil A (2012) Biohydrometallurgy techniques of low grade ores: a review on black shale. Hydrometallurgy 117–118:1–12
- 9. Suzuki I (2001) Microbial leaching of metals from sulfide minerals. Biotechnol Adv 19(2):119–132
- Bosecker K (1997) Bioleaching: metal solubilization by microorganisms. FEMS Microbiol Rev 20:591–604
- Choi M-S, Cho K-S, Kim D-S, Ryu H-W (2005) Bioleaching of uranium from low grade black schists by *Acidithiobacillus ferrooxidans*. World J Microbiol Biotechnol 21(3):377–380
- Renman R, Jiankang W, Jinghe C (2006) Bacterial heap-leaching: practice in Zijinshan copper mine. Hydrometallurgy 83(1-4):77–82
- Panda S, Sanjay K, Sukla LB, Pradhan N, Subbaiah T, Mishra BK, Prasad MSR, Ray SK (2012) Insights into heap bioleaching of low grade chalcopyrite ores: a pilot scale study. Hydrometallurgy 125–126:157–165
- Fu B, Zhou H, Zhang R, Qiu G (2008) Bioleaching of chalcopyrite by pure and mixed cultures of *Acidithiobacillus* spp. and *Leptospirillum ferriphilum*. Int Biodeterior Biodegrad 62(2):109–115
- de Souza AD, Pina PS, Leão VA (2007) Bioleaching and chemical leaching as an integrated process in the zinc industry. Miner Eng 20(6):591–599
- Ke Y, Peng N, Xue K, Min X, Chai L, Pan Q, Liang Y, Xiao R, Wang Y, Tang C, Liu H (2018) Sulfidation behavior and mechanism of zinc silicate roasted with pyrite. Appl Surf Sci 435:1011–1019
- Schippers A, Hedrich S, Vasters J, Drobe M, Sand W, Willscher S (2014) Biomining: metal recovery from ores with microorganisms. In: Schippers A, Glombitza F, Sand W (eds) Geobiotechnology I, vol 141. Advances in biochemical engineering/ biotechnology. Springer, Berlin, pp 1–47
- Wang X, Liu Y, Sun Z, Li J, Chai L, Min X, Guo Y, Li P, Zhou Z (2017) Heap bioleaching of uranium from low-grade granite-type

ore by mixed acidophilic microbes. J Radioanal Nucl Chem 314(1):251-258

- Umanskii A, Klyushnikov A (2013) Bioleaching of low grade uranium ore containing pyrite using *A. ferrooxidans* and *A. thiooxidans*. J Radioanal Nucl Chem 295(1):151–156
- Rashidi A, Roosta-Azad R, Safdari SJ (2014) Optimization of operating parameters and rate of uranium bioleaching from a lowgrade ore. J Radioanal Nucl Chem 301(2):341–350
- 21. Li Q, Sun J, Ding D, Wang Q, Shi W, Hu E, Wang X, Jiang X (2017) Characterization and uranium bioleaching performance of mixed iron- and sulfur-oxidizers versus iron-oxidizers. J Radioanal Nucl Chem 314(3):1939–1946
- Gan M, Jie S, Li M, Zhu J, Liu X (2015) Bioleaching of multiple metals from contaminated sediment by moderate thermophiles. Mar Pollut Bull 97(1–2):47–55
- Gan M, Zhou S, Li M, Zhu J, Liu X, Chai L (2015) Bioleaching of multiple heavy metals from contaminated sediment by mesophile consortium. Environ Sci Pollut Res Int 22(8):5807–5816
- 24. Xiang Y, Wu P, Zhu N, Zhang T, Liu W, Wu J, Li P (2010) Bioleaching of copper from waste printed circuit boards by bacterial consortium enriched from acid mine drainage. J Hazard Mater 184(1–3):812–818
- 25. Chen S, Yang Y, Liu C, Dong F, Liu B (2015) Column bioleaching copper and its kinetics of waste printed circuit boards (WPCBs) by *Acidithiobacillus ferrooxidans*. Chemosphere 141:162–168
- Park YJ, Fray DJ (2009) Recovery of high purity precious metals from printed circuit boards. J Hazard Mater 164(2–3):1152–1158
- 27. Qiu G, Li Q, Yu R, Sun Z, Liu Y, Chen M, Yin H, Zhang Y, Liang Y, Xu L, Sun L, Liu X (2011) Column bioleaching of uranium embedded in granite porphyry by a mesophilic acidophilic consortium. Bioresour Technol 102(7):4697–4702
- Liu DG, Min XB, Ke Y, Chai LY, Liang YJ, Li YC, Yao LW, Wang ZB (2018) Co-treatment of flotation waste, neutralization sludge, and arsenic-containing gypsum sludge from copper smelting: solidification/stabilization of arsenic and heavy metals with minimal cement clinker. Environ Sci Pollut Res Int 25(8):7600–7607
- Abhilash SS, Mehta KD, Kumar V, Pandey BD, Pandey VM (2009) Dissolution of uranium from silicate-apatite ore by *Acidithiobacillus ferrooxidans*. Hydrometallurgy 95(1–2):70–75
- Muñoz JA, Ballester A, González F, Blázquez ML (1995) A study of the bioleaching of a Spanish uranium ore. Part II: orbital shaker experiments. Hydrometallurgy 38(1):59–78
- Pal S, Pradhan D, Das T, Sukla LB, Chaudhury GR (2010) Bioleaching of low-grade uranium ore using *Acidithiobacillus ferrooxidans*. Indian J Microbiol 50(1):70–75
- 32. H-d P, H-y Y, L-l T, C-b Z, Y-s Z (2012) Control method of chalcopyrite passivation in bioleaching. Trans Nonferr Met Soc China 22(9):2255–2260
- Xiong X-x G, G-h BJ-r, S-k L (2015) Bioleaching and electrochemical property of marmatite by Sulfobacillus thermosulfidooxidans. Trans Nonferr Met Soc China 25(9):3103–3110