



Eclética Química

ISSN: 0100-4670

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Universidade Estadual Paulista Júlio de
Mesquita Filho
Brasil

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Eclética Química, vol. 42, 2017, pp. 40-50

Universidade Estadual Paulista Júlio de Mesquita Filho
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Effect of redox potential on chalcopyrite dissolution imposed by addition of ferrous ions

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ARTICLE INFO

Article history:

Received: August 28, 2017

Accepted: November 16, 2017

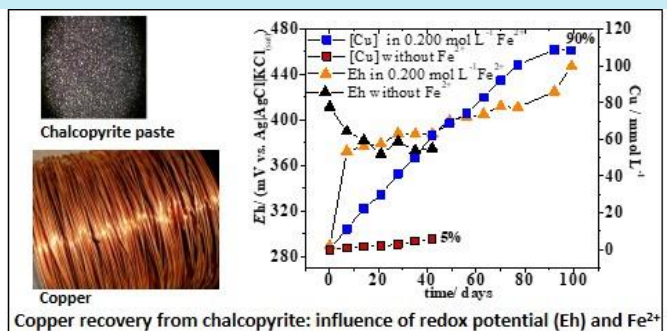
Published: December 30, 2017

Keywords:

1. bioleaching
2. chalcopyrite
3. redox potential
4. ferrous ions

ABSTRACT: Copper is one of the metals with great economic interest and about 70% of it is found in nature as chalcopyrite mineral (CuFeS_2). Due to its slow dissolution kinetics, chalcopyrite dissolution is still a challenge for industries and researchers. Control of redox potential though has been said to be the key for increasing dissolution rate in chalcopyrite leaching systems. The current work investigated the effect of redox potential by ferrous ions addition on chalcopyrite (bio)leaching experiments. In abiotic systems, 90% of copper was leached into solution when redox potential was kept low (around 420 mV/Ag/AgCl) whereas in bacterial systems (610 mV/Ag/AgCl) low copper dissolution was observed. SEM analysis suggested presence of new phases that were confirmed by XRD to be elemental sulfur and jarosites and a significant decrease of chalcopyrite peak heights in these analyses was also observed. Jarosite was the only new phase detected in bacterial systems. Passivation of chalcopyrite surface did not occur in either conditions. The results clearly show that low redox potential along with presence of ferrous ions has positively influenced copper recovery, confirming the existence of a critical redox potential range where chalcopyrite dissolution is favored.

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Copper recovery from chalcopyrite: influence of redox potential (Eh) and Fe^{2+}

1. Introduction

Chalcopyrite is the primary copper mineral for production of copper metal. It is known to be one of the most stable copper minerals due to its structurally face-centered tetragonal lattice and for responding poorly to hydrometallurgical processes when compared to copper oxides ores^{1,2}.

Slow dissolution rates of chalcopyrite have been attributed to a possible blockage by a superficial layer whose formation and action mechanisms on chalcopyrite are still quite contradictory in the

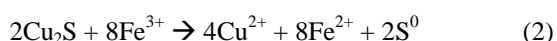
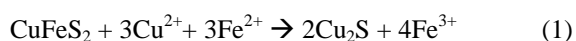
literature. Córdoba *et al.*³ associated this layer formation with high concentration of ferric ion, consequently, high redox potential values, potentiating jarosite precipitates formation ($\text{XFe}(\text{SO}_4)_2(\text{OH})_6$ where $\text{X} = \text{K}^+, \text{NH}_4^+, \text{Na}^+$ or H_3O^+) and ultimately blocking chalcopyrite surface. Studies have already discussed that the formation of a compact sulfur layer, iron-bearing precipitates or jarosites has been responsible for covering chalcopyrite surface and leading to a slow and/or incomplete copper extraction^{4,5,6}. It is noteworthy that the formation of jarosites is favored in media with high values of redox potential, such as

bioleaching medium, in which the production of ferric ions is intense due to bacterial activity. This layer has also been described as a copper polysulfide layer, CuS_n (where $n > 2$), formed during initial stages of chalcopyrite oxidation⁴. However, Klauber⁵ has claimed the characterization techniques used to identify the possible layer formed on chalcopyrite surface and stated that the only products resulting from mineral oxidation that had been demonstrated were sulfur and jarosites. The author also stated that sulfur is not able to stop chalcopyrite dissolution since this element has a relative mobility, but the formation of jarosites seems to hinder chalcopyrite dissolution.

In order to accelerate dissolution rates, jarosites precipitation must be prevented. One of the solutions discussed to overcome this issue considers the control of redox potential throughout (bio)leaching processes that could be achieved either electrochemically by applying a redox potential from an external source through a working electrode or chemically by the addition of reducing and oxidizing agents⁷.

The maintenance of the redox potential in a narrow range is discussed by Hiroyoshi and colleagues^{9,10}. During the chemical leaching of chalcopyrite in ferric sulfate media, both rate and yield of copper dissolution is maximum within narrow range of redox potential around 400–450 mV/ $\text{Ag}|\text{AgCl}|\text{KCl}_{(\text{sat})}$, whereas redox potential values above this range the blockage of mineral surface does occur mainly due to jarosite precipitation. The authors described a greater metal recovery in solutions with different ferrous, ferric and cupric ions concentrations⁹. The presence of certain concentrations of these ions in solution allowed the maintenance of an optimum redox potential for copper recovery due to formation of less refractory copper intermediates than chalcopyrite, such as chalcocite (Cu_2S).

The maintenance of a certain $\text{Fe}^{3+}/\text{Fe}^{2+}$ ratio can increase the copper extraction rate by favoring the following reactions (Equation 1 and 2) occurring in a chalcopyrite massive electrode¹⁰:



Such reactions occur in a narrow potential range (400–430 mV/ $\text{Ag}|\text{AgCl}|\text{KCl}_{(\text{sat})}$) and chalcocite formation maintain chalcopyrite indirect oxidation and

ferric ions concentration is insufficient for jarosite precipitation.

A bioleaching investigation including microscopy and electrochemistry showed that the presence of chalcocite significantly enhanced copper extraction and formation of jarosites seems to be insufficient for blocking the surface since these compounds are highly porous and easily lost from the surface¹¹.

From the studies presented it is possible to note that the redox potential control apparently can improve the copper dissolution regardless of the presence or absence of precipitates.

In this context, the aim of the current work was to understand the role of redox potential in bioleaching processes by addition of different concentration of ferrous ions. It is important to emphasize that the concentrations used in this work were higher than that found in the previous literature due to our interest in study the effects of the potential in a larger range that best reflects practical conditions.

2. Experimental Section

2.1. Mineral sample

The chalcopyrite sample used in this study was obtained from La Chorrera (Cisneros, Colombia), kindly provided by Prof. Marco Antonio Marquez. The mineral was ground to 100% <115 mesh (125 μm opening). The metal content was 30.86% Cu and 24.4% Fe by acid digestion followed by atomic absorption analysis. The X-ray diffraction (XRD) revealed the presence of chalcopyrite as only sulfide phase (Figure 1).

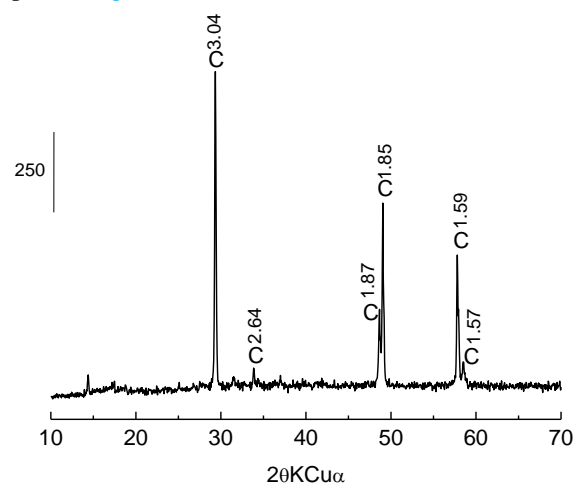


Figure 1. X-ray diffractogram of untreated chalcopyrite sample. The vertical bar shows the relative scale of counts. The d-values are given in Ångströms.

2.2. Bacteria and growth conditions

Acidithiobacillus ferrooxidans, strain LR¹² was used in this study. The culture was grown in mineral salts medium (MSM) which contained (per liter) 0.5 g (NH₄)₂SO₄, 0.5 g K₂HPO₄, and 0.5 g MgSO₄·7H₂O at pH 1.80 adjusted with sulfuric acid. *At. ferrooxidans* LR stock culture was grown using ferrous ions (0.120 mol L⁻¹ FeSO₄·7H₂O), as energy source, quantity enough to increase cell yields to 10⁸ cells mL⁻¹¹³. The mineral salts medium and the mineral sample were autoclaved at 121 °C by 20 min whereas ferrous sulfate was filtrated using 0.45 µm membrane filter (Millipore). Aseptic techniques were used in handling, inoculation and sampling of the cultures. Cultures were incubated at 30 °C in an orbital shaker at 150 rpm. Cell counts were not determined in the experiments due to the uncertainty of distribution of attached and planktonic cells during chalcopirite oxidation. As an approximation, the inoculation of Fe²⁺-grown *At. ferrooxidans* yielded initial concentrations of the low range of 10⁸ cells mL⁻¹ in chalcopirite cultures in shake flasks.

2.3. Acid Digestion

The mineral sample was dried at 60 °C until constant weight. Then, 80.0 mL of a concentrated hydrochloric acid and nitric acid mixture (in a proportion of 3:1) was added to 2.00 g of the mineral sample that remained reacting for 24 hours. This mixture was heated in a hot plate and 10.0 mL of concentrated hydrochloric acid were added successively until substantial decrease of volume. Hydrochloric acid 50% was added to this mixture to a final volume of 30.0 mL and then was filtrated using 0.45 µm membrane filter (Millipore). The metal-containing solution was analyzed by atomic absorption spectroscopy for copper and iron quantification.

2.4. Leaching experiments

All leaching tests were performed in an orbital shaker at 150 rpm and 30 °C using Erlenmeyer flasks containing mineral salts medium¹³ at pH 1.80 plus ferrous sulfate as energy source in different concentrations (0, 0.100, 0.200 and 0.300 mol L⁻¹) and 2.50% (w/v) of chalcopirite sample. Biotic conditions were performed using 5.00% (v/v) of *At. ferrooxidans*

LR. The experiments were carried out in duplicate with a final volume of 150 mL. Evaporation was compensated for adding sterile distilled water.

2.5. Analytical procedures

Shake flasks were periodically sampled (3.00 mL) in order to obtain kinetic information on metal dissolution besides pH and redox potential measured against Ag|AgCl|KCl_(sat) reference electrode. For chemical analysis, samples were centrifuged (10,000 g for 15 min). Aliquots of supernatants were preserved in HNO₃ 2.0% for analysis of dissolved copper and total iron by atomic absorption spectroscopy.

Ferrous ion quantification was performed using a photocolometric method based on the formation of a reddish colored complex of Fe²⁺ with ortho-phenantroline. The solution was analyzed in an UV-Vis spectrophotometer at a wavelength of 510 nm¹⁴.

Finally, solid residues were characterized by XRD using the SIEMENS D5000 diffractometer equipped with a diffracted-beam monochromator and CuKα radiation. Samples were scanned from 10 to 70° 2θ at 0.05° 2θ increments with 2.0 s counting time.

SEM-EDS analysis were carried out only in leaching final residues using a Scanning Electron Microscopy, JSM – 7500F, and X ray microanalysis, Scientific – Noran System.

3. Results and discussion

3.1. Leaching experiments

pH, redox potential, ferrous ions and total iron concentrations were monitored throughout the experiment. The initial pH was set up to 1.80, however pH values ranged from 1.70 to 2.50 (Figure 2). In cases of pH higher than 2.00, it was necessary to adjust it to 1.80 with H₂SO₄ (50%) in order to avoid precipitate formation, such as jarosites, which is more auspicious in pH values higher than 2.00 and their formation could be unfavorable for metal recovery. In abiotic conditions, pH adjustments were necessary during the experiment time whereas in bacterial conditions these adjustments were required just at the beginning. Inoculated condition determined an oxidant environment that reflected in all monitored parameters.

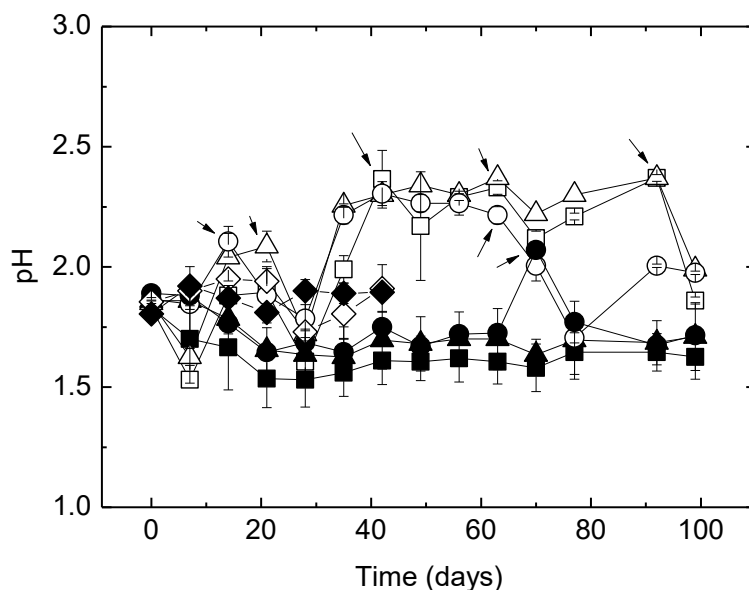


Figure 2. Changes in pH values on chalcopyrite leaching using different concentrations of ferrous ions in absence (open symbols) and presence (solid symbols) of bacteria. Rhombus: 0 mol L⁻¹; square: 0.100 mol L⁻¹; triangle: 0.200 mol L⁻¹; and circle: 0.300 mol L⁻¹ of ferrous ions. Black arrows indicate pH adjustment.

Figure 3 illustrates the ferrous ions consumption. In abiotic conditions ferrous ions concentration was maintained stable until the end of experiment due to the absence of an agent in solution capable of oxidizing ferrous ions in the same rate as the bacterium *At. ferrooxidans*. On the contrary, in inoculated conditions ferrous ions present in solution were oxidized to ferric ions within 7 days showing the bacterial capability of obtaining energy from ferrous ions and generating an extremely oxidant environment.

After 60 days a decrease in ferrous ions concentration can be clearly seen in abiotic conditions containing initial concentrations of 0.200 and 0.300 mol L⁻¹ of Fe²⁺ (Figure 3). This occurrence can be explained by the ferrous ions oxidation by oxygen and/or the consumption of these ions caused by chalcopyrite reduction reaction in the presence of Cu²⁺ and H⁺⁹.

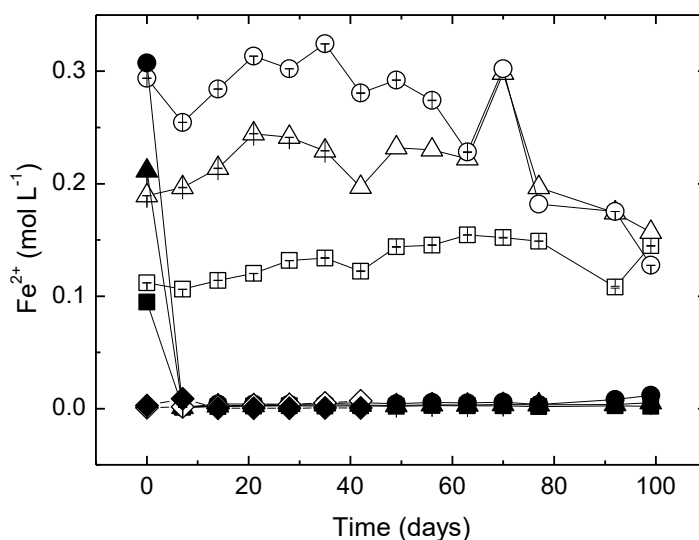


Figure 3. Changes in ferrous ions concentration on chalcopyrite leaching using different concentrations of ferrous ions in absence (open symbols) and presence (solid symbols) of bacteria. Rhombus: 0 mol L⁻¹; square: 0.100 mol L⁻¹; triangle: 0.200 mol L⁻¹; and circle: 0.300 mol L⁻¹ of ferrous ions.

In all abiotic conditions was observed a slight increase in total soluble iron concentration (Fe^{2+} and Fe^{3+}) throughout the experiment, being more pronounced in the conditions containing 0.200 and 0.300 mol L⁻¹ (data not shown) possibly due to chalcopyrite oxidation and consequent release of iron from its matrix. Under biotic conditions no significant change in total soluble iron concentration was observed in both 0.100 and 0.200 mol L⁻¹ Fe^{2+} conditions. However, total soluble iron concentration decreased during the experiment in the 0.300 mol L⁻¹ Fe^{2+}

condition mainly due to bacterial activity which accelerates the conversion of ferrous into ferric ions through a proton-consuming reaction, therefore increasing pH, favoring jarosites formation^{2,15}.

Redox potential was also monitored throughout the experiment as presented in Figure 4. In abiotic conditions the redox potential values ranged from 300 to 470 mV whereas in bacterial conditions the redox potential increased progressively and after 7 days of experiment it reached around 600 mV and this value was maintained until the end.

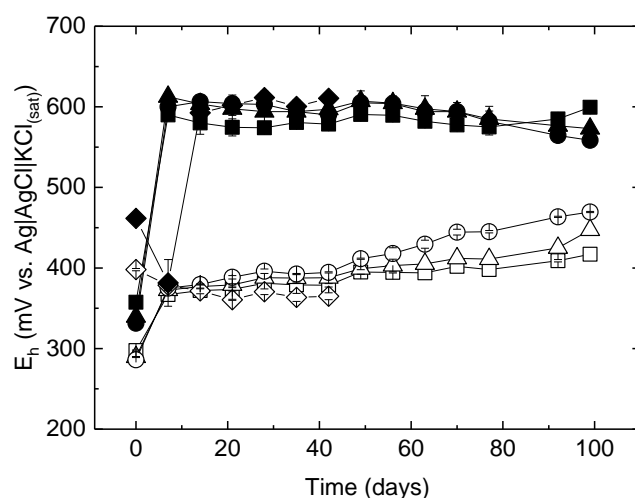


Figure 4. Redox potential measurements on chalcopyrite leaching using different concentrations of ferrous ions in absence (open symbols) and presence (solid symbols) of bacteria: rhombus- 0 mol L⁻¹; square- 0.100 mol L⁻¹; triangle- 0.200 mol L⁻¹; and circle- 0.300 mol L⁻¹ of ferrous ions.

Figure 5 shows the copper recovery in abiotic and bacterial conditions.

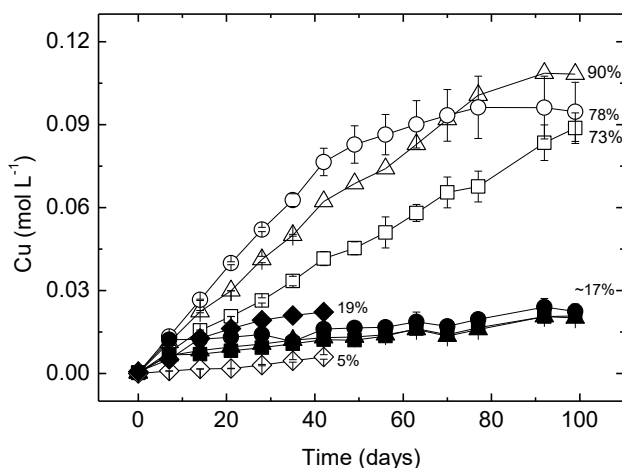


Figure 5. Time course of copper dissolution. Abiotic and bacterial conditions are represented by open and filled symbols, respectively. Rhombus: 0 mol L⁻¹; square: 0.100 mol L⁻¹; triangle: 0.200 mol L⁻¹; and circle: 0.300 mol L⁻¹ of ferrous ions.

In this study the difference in ferrous ions concentrations in abiotic conditions was crucial to determine soluble copper concentration. The greatest copper recovery was observed in the $0.200 \text{ mol L}^{-1} \text{ Fe}^{2+}$ condition, approximately 90% copper recovered after 100 days. For each abiotic condition tested (0, 0.100, 0.200 and 0.300 mol L^{-1} of ferrous ions), the final redox potential was respectively 365, 410, 435 and 470 mV, leaching about 5, 70, 90 and 80% copper, respectively. Different concentrations of ferrous ions in chalcopyrite leaching experiments resulted in different amount of copper solubilized as demonstrated previously¹⁶ for different mineral and conditions, confirming the effect of ferrous ions concentration in chalcopyrite dissolution. The final solid residue of the $0.200 \text{ mol L}^{-1} \text{ Fe}^{2+}$ condition was used to perform an acid digestion in order to compare with the results obtained from the liquid phase and the copper recovery obtained using this method was over 99%. Both 0.100 and $0.200 \text{ mol L}^{-1} \text{ Fe}^{2+}$ conditions presented a very similar trend regarding redox potential throughout the experiment whereas in the $0.300 \text{ mol L}^{-1} \text{ Fe}^{2+}$ condition the redox potential starts to increase gradually after 60 days.

Copper dissolution rate decreased slightly when the redox potential reached about 420 mV after 90 days in the $0.200 \text{ mol L}^{-1} \text{ Fe}^{2+}$ condition and after 50 days in the $0.300 \text{ mol L}^{-1} \text{ Fe}^{2+}$ until stops completely (Figures 4 and 5). This incident could not be seen in $0.100 \text{ mol L}^{-1} \text{ Fe}^{2+}$ because the redox potential values did not reach 420 mV even after 100 days of experiment.

Chalcopyrite bioleaching in presence of *At. ferrooxidans* LR (Figure 5) resulted in a slight increase of copper dissolution rate at the beginning of experiment, followed by a decrease which was observed after 7 days and it was kept almost constant until the end of the experiment. Only 11% of copper was released into solution after 40 days in all bacterial conditions studied and even after 100 days of experiment copper recovery was not higher than 17%. Despite the difference in ferrous ions concentration added to the flasks (0.100 , 0.200 and 0.300 mol L^{-1}),

the final copper recovery rate obtained was almost the same value. This result was confirmed by EIS measurements for shorter immersion times in other studies^{17,18}, where the chalcopyrite surface seems to be partially blocked indicated by an increase of real impedance in the presence of *At. ferrooxidans*.

Results showed that the presence of bacteria caused a detrimental effect on copper dissolution in all concentrations studied in comparison with abiotic conditions. Low copper recovery can be associated to the high initial ferrous ions concentration readily available in solution, since bacteria prefer to oxidize soluble ferrous ions than chalcopyrite as energy source, as described by Gómez *et al.*¹⁹. Oxidizing chalcopyrite rather than soluble ferrous iron may favor biofilm formation that could hinder further chalcopyrite dissolution as described elsewhere^{17,18}.

Li *et al.*²⁰ stated that the use of microbes for leaching minerals such as chalcocite, covellite and pyrite showed a beneficial effect on metal recovery; however this effect has been under discussion concerning chalcopyrite. On the other hand, some reviews have shown that there is a general acceptance that bacterial leaching of chalcopyrite does have a beneficial effect especially using high temperature and moderate thermophiles and thermophiles^{1,2}.

Third *et al.*²¹ carried out a series of bioleaching and chemical leaching experiments in which they concluded that the redox potential is more important for determining chalcopyrite leaching rates than bacterial activity. They also concluded that high initial ferric ions concentration and high redox potential were detrimental to chalcopyrite leaching rates whereas Fe^{2+} -containing leaching medium raised chalcopyrite dissolution significantly.

The final residues of leaching experiments were analyzed by SEM/EDS and XRD. SEM analyses showed new crystalline phases other than chalcopyrite in solid residues. Analyses using EDS suggested them to be jarosite and elemental sulfur (Figure 6) that were confirmed by XRD analyses (Figure 7).

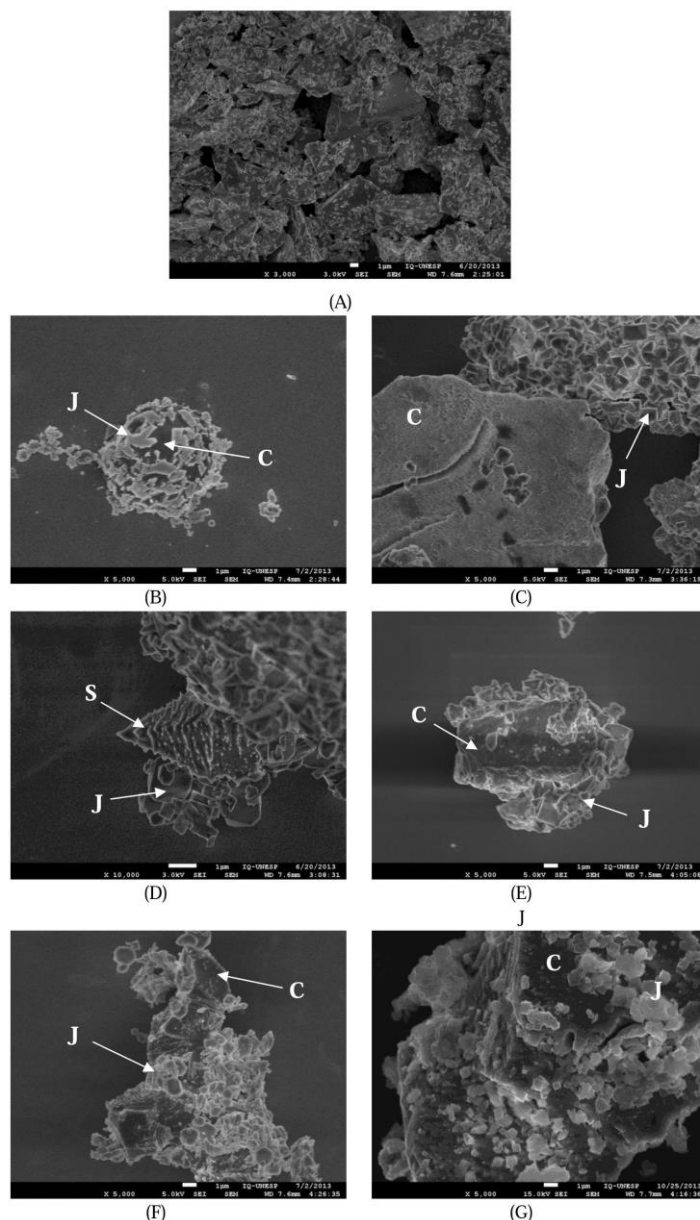


Figure 6. General outline of SEM analyses of solid residues from chalcopyrite (bio)leaching after 100 days of experiment. Untreated chalcopyrite (A); addition of $0.100 \text{ mol L}^{-1} \text{Fe}^{2+}$: abiotic (B) and bacterial (C) conditions; addition of $0.200 \text{ mol L}^{-1} \text{Fe}^{2+}$: abiotic (D) and bacterial (E) conditions and addition of $0.300 \text{ mol L}^{-1} \text{Fe}^{2+}$: abiotic (F) and bacterial (G) conditions. Letter designations: C = chalcopyrite, J = jarosite and S = Sulfur.

Figure 7 shows X-ray patterns of an untreated chalcopyrite sample and solid residues after 100 days of leaching with addition of ferrous ions. The diffractograms showed the presence of jarosites and elemental sulfur, as cumulative phases, since sulfur formation was first observed after 14 days of experiment. It is noteworthy that the intensity of the

chalcopyrite's peak has a significant decrease which confirms the percentage of recovery obtained in the leachate. In all bacterial conditions only jarosite was formed during the experiment (data not shown). In the control condition (without addition of ferrous ions) formation of a new crystalline phase was not observed.

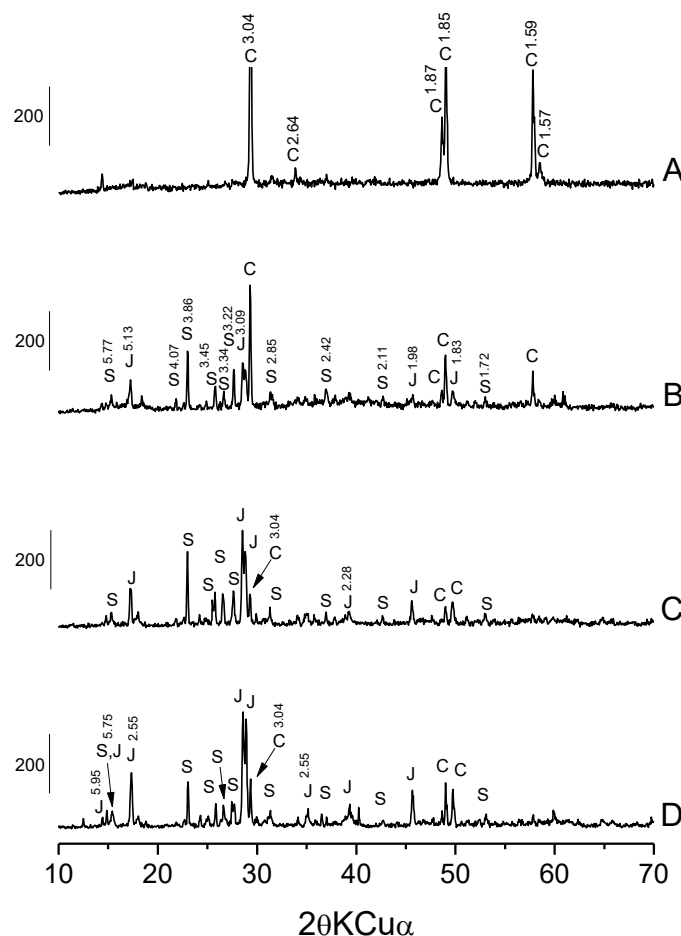


Figure 7. X-ray diffractograms of solid residues from chalcopirite bioleaching after 100 days of experiment in abiotic conditions. A: Untreated chalcopirite, B: addition of 0.100 mol L⁻¹, C: 0.200 mol L⁻¹ and D: 0.300 mol L⁻¹ of ferrous ions. Letter designations: C = chalcopirite, J = jarosite and S = sulfur. The vertical bar shows the relative scale of counts. The d-values are given in Ångströms.

Previously, it has been reported that the precipitation of elemental sulfur can hinder chalcopirite leaching as well as iron precipitates^{22,23}. Despite the formation of elemental sulfur and its progressive accumulation, detrimental effects on chalcopirite dissolution were not observed. On the contrary, the 0.200 mol L⁻¹ Fe²⁺ condition was able to recover over 99% of copper.

Considering the results obtained in this present work, the effect of continuous chalcopirite dissolution can be attributed to the maintenance low redox potential as well as the presence of ferrous ions in solution. Several authors have described the positive effect of both redox potential and ferrous ions on chalcopirite dissolution, however attributing different significance of each one^{7,9,10,21,24,25,26}. For instance, Third *et al.*²¹ considered the redox potential a parameter more significant to determine chalcopirite leaching rates than bacterial cell number or their

activity. The authors also stated that addition of ferrous ions to the systems increased copper dissolution rate in 2.7 fold than when the same quantity of ferric ions was added. Several other authors^{24,25,26} have focused only on a critical redox potential range (400-450 mV) over which chalcopirite (bio)leaching is hindered.

This current study showed that the control condition tested (without addition of ferrous ions and without bacteria) was the system that reached the lowest redox potential value and the lowest copper recovery. Therefore, in an optimized system, it is necessary not only low redox potential values (400-450 mV) but also presence of ferrous ions to ensure higher rates of copper recovery. Recently, it was demonstrated by electrochemical impedance spectroscopy that ferrous ions activated chalcopirite surface even in the presence of surface precipitates and the redox potential range could help the maintenance of this activation¹⁸. Thus, we propose a new approach to interpret the effect

of ferrous ions and redox potential on chalcopyrite dissolution considering the combined effect of these

parameters, since both parameters cannot be interpreted separately (Figure 8).

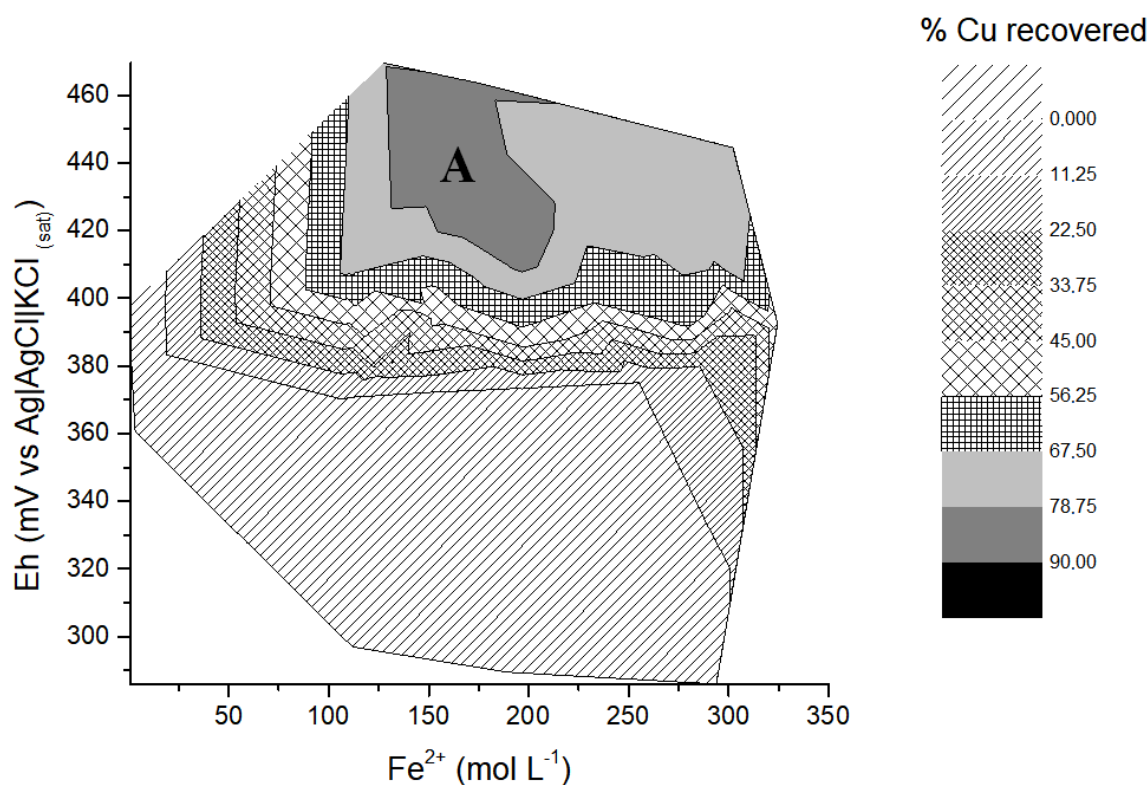


Figure 8. Combined effect of ferrous iron concentration and redox potential on chalcopyrite dissolution under abiotic condition.

Maximum copper recovery is delimited by the region defined by the dark grey color (region A) where the redox potential interval (410-470 mV) is associated with ferrous ion concentration between 0.125-0.215 mol L⁻¹. The presence of bacteria generates an oxidizing atmosphere, which has an unfavorable effect in copper dissolution due to the high redox potential in these conditions and the high ferric ions concentration in solution.

4. Conclusions

The dissolution of chalcopyrite is favored by a simultaneous effect of ferrous ion concentration and redox potential interval: 0.125-0.215 mol L⁻¹ Fe²⁺ and 410-470 mV. Iron precipitates and elemental sulfur did not affect copper dissolution in any abiotic conditions studied.

Presence of ferrous ions in abiotic systems enhanced significantly chalcopyrite dissolution

recovering over 99% of copper. High dissolution rates in chalcopyrite leaching are related to the maintenance of low redox potential as well as the presence of ferrous ions in solution and both effects must be considered together.

5. Acknowledgments

The authors gratefully acknowledge financial support and scholarships from the Brazilian funding agencies: CNPq (ALAS for master fellowship), Additional support for this study was received from CAPES (ALA) and FAPESP (FAA proc. no 2010/13089-1 and DB, 19868-5/2011).

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