Leaching of Pure Chalcocite with Reject Brine and MnO₂ from Manganese Nodules

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Abstract: Chalcocite (Cu₂S) has the fastest kinetics of dissolution of Cu in chlorinated media of all copper sulfide minerals. Chalcocite has been identified as having economic interest due to its abundance, although the water necessary for its dissolution is scarce in many regions. In this work, the replacement of fresh water by sea water or by reject brine with high chloride content from desalination plants is analyzed. Additionally, the effect of adding MnO₂ from available manganese nodules in vast quantities at the bottom of the sea is studied. Reject brine shows better results than sea water, and the addition of MnO₂ to the brine significantly increases the kinetics of chalcocite dissolution in a short time. H₂SO₄ concentration is found to be irrelevant when working at high concentrations of chloride and MnO₂. The best results, 71% Cu extractions in 48 h, are obtained for reject brine, 100 mg of MnO₂ per 200 g of mineral and H₂SO₄ 0.5 mol/L. The results are expected to contribute to a sustainable process of dissolution of chalcocite by using the reject brine from desalination plants.

Keywords: sulfide leaching; chalcocite dissolution; desalination; reject brine; replace water; sustainability

1. Introduction

Currently, 19.7 million tons of copper are produced worldwide [1,2], mainly by pyrometallurgical processes (75%) and, to a lesser extent, by hydrometallurgy (25%) [3,4].

The vast majority of copper minerals in the world correspond to sulfide minerals and a smaller quantity to oxidized minerals [5–8]. Among the sulfurous minerals, the most abundant copper mineral is chalcopyrite [9–11], followed by chalcocite [12]. However, chalcopyrite is a very refractory mineral to be treated hydrometallurgically [13]. On the other hand, chalcocite has been shown to be relatively easy to treat using hydrometallurgical processes [14]. The literature reports an important number of works on the dissolution of chalcocite in acidic media, pressure leaching [15], bioleaching [15–17] and
leaching in chlorinated media [18–21]. This last process is very attractive because it is cost effective and leads to high copper extractions.

There is no record in the literature on the use of MnO$_2$ as an oxidizing agent in the leaching of chalcocite in chlorinated media. However, there are records in which this oxidant is added to improve the dissolution of chalcopyrite with good results [22–24]. Havlík et al. [24] use HCl and manganese nodules to extract Cu from chalcopyrite. This study finds that, when working at high concentrations of MnO$_2$ (MnO$_2$/CuFeS$_2$ ratio of 4/1) and HCl (4 mol/L), the best copper extractions are obtained, reaching values above 40% at room temperature and near 70% at 50 °C.

The dissolution of chalcocite in chlorinated media has been analyzed several times [14,20,21,25,26], obtaining high copper extraction in the presence of oxidants, such as ferric or cupric, at high temperature, which greatly improves the leaching kinetics of this mineral. To evaluate the effect of chloride, both the concentration and the source have been analyzed, that is, sea water, HCl, NaCl, FeCl$_3$, etc. Cheng and Lawson [25] have proposed that chloride acts as a catalyst that forms long sulfur crystals that allow the diffusion of oxidizing agents within the mineral avoiding the formation of a passivating layer that retards the dissolution of the mineral. Several authors [27,28] have obtained similar results, and to demonstrate the mechanism they have measured the porosity at the mineral surface when copper sulfides are leached in the presence of NaCl, HCl, FeCl$_3$ or other sources of chloride. It is this porosity that allows the mineral to contact the leaching agent through these sulfur layers that form on the mineral surface.

Two reactions are proposed for the dissolution of chalcocite in chlorinated media and in the presence of manganese (IV) as an oxidizing agent.

\[
2 \text{Cu}_2\text{S} + \text{MnO}_2 + 4 \text{Cl}^- + 4\text{H}^+ = 2 \text{CuCl}_2^- + \text{Mn}^{2+} + 2 \text{H}_2\text{O} + 2 \text{CuS} \ \Delta G^0 = -138.59 \text{ kJ} \tag{1}
\]

\[
2 \text{CuS} + \text{MnO}_2 + 4 \text{Cl}^- + 4\text{H}^+ = 2 \text{CuCl}_2^- + \text{Mn}^{2+} + 2 \text{H}_2\text{O} + 2 \text{S} \ \Delta G^0 = -84.512 \text{ kJ} \tag{2}
\]

In the first leaching stage (Equation (1)), chalcocite is converted to covellite; this reaction is thermodynamically possible according to the Gibbs free energy that is negative under the described conditions. The second reaction (Equation (2)) is slower. The energies were calculated using the HSC 5.1 software. Thus, Equation (1) is more likely to occur than Equation (2) as suggested by other authors [12,29].

The shortage of fresh water in various regions of the world is a major economic, environmental and social problem [30]. The use of sea water has become increasingly important for mining, for example, in northern Chile, not only for its positive effects on leaching processes due to its chloride content, but as a strategic and indispensable resource for the sustainability of the industry [4,31,32]. A very attractive alternative is the use of wastewater from desalination plants, which, following the example of Chile, are abundant in the north of the country. Desalination plants produce drinking water for the population, but the reject brine pollutes the sea, threatening the marine ecosystem, and therefore alternatives are required to recycle or reuse this concentrated brine.

In this study, the kinetics of chalcocite dissolution in acid solution and at room temperature in chlorinated media and in the presence of manganese nodules are determined. The kinetics of chalcocite dissolution in sea water and in reject brine from a desalination plant are compared, and the effect of MnO$_2$ and sulfuric acid on the percentage of copper extraction is evaluated, at short and long times.

2. Materials and Methods

2.1. Chalcocite

The chalcocite mineral sample was obtained from Mina Atómica, located in Antofagasta, Chile. The material was reduced to reach a size range between ~150 and ~106 µm. The grinding was done in a porcelain mortar to avoid contamination. The chemical composition was determined by inductively coupled plasma atomic emission spectrometry (ICP-AES, FEI Company, Brisbane,
Australia). Table 1 shows the results. In addition, the mineralogy of the sample was analyzed using a Bruker X-ray diffractometer (Bruker, Billerica, MA, USA), automatic and computerized model D8, Figure 1 shows the results. The sample was 99.90% chalcocite.

Table 1. Chemical analysis of the chalcocite ore sample.

| Component | Cu   | S   |
|-----------|------|-----|
| Mass (%)  | 79.83| 20.17|

2.2. Manganese Nodules

The MnO$_2$ used came from manganese nodules collected during the 1970s from the Blake Plateau in the Atlantic Ocean. The sample was reduced in size in a porcelain mortar until reaching a size range between −140 and +100 µm. The chemical composition was determined by inductively coupled plasma atomic emission spectrometry (ICP-AES). Table 2 shows the results. The mineralogy of the sample was analyzed using Bruker® M4-Tornado μ-XRF table-top equipment (Fremont, CA, USA). μ-XRF data interpretation showed the sample is comprised of a pre-existent nodule fragment forming the core with concentric layers precipitated around it at later stages. The experiments showed pyrolusite (MnO$_2$) as the predominant phase (See Table 3).

Table 2. Chemical analysis of manganese ore.

| Component | Mn  | Fe  | Cu  | Co  |
|-----------|-----|-----|-----|-----|
| Mass %    | 15.96| 0.45| 0.12| 0.29|

Table 3. Mineralogical analysis of manganese ore.

| Component | MgO | Al$_2$O$_3$ | SiO$_2$ | P$_2$O$_5$ | SO$_3$ | K$_2$O | CaO | TiO$_2$ | MnO$_2$ | Fe$_2$O$_3$ |
|-----------|-----|-------------|---------|------------|--------|--------|-----|---------|---------|-------------|
| Mass (%)  | 3.54| 3.69        | 2.97    | 7.20       | 1.17   | 0.33   | 22.48| 1.07    | 29.85   | 26.02       |
2.3. Reagents and Leaching Tests

Sea water from Antofagasta (Chile) and reject brine from Aguas Antofagasta, Antofagasta (Chile), were used. The sulfuric acid used in the leaching tests is Merck, grade p.a., purity 95–97%, density 1.84 kg/L and molecular weight 98.08 g/mol. Leaching tests were carried out in a 50 mL glass reactor with a 0.01 solid to liquid (S/L) ratio of leaching solution. A total of 200 mg of chalcocite ore was kept suspended by agitation with the use of a 5-position magnetic stirrer (IKA ROS, CEP 13087-534, Campinas, Brazil) at a speed of 600 rpm. The tests were conducted at room temperature of 25 °C at different concentrations of sulfuric acid and chloride, and leaching times. The tests were performed in duplicate. Analyses were performed on 5 mL undiluted samples using atomic absorption spectrometry with a coefficient of variation ≤5% and a relative error between 5% and 10%. Measurements of pH and oxidation-reduction potential (ORP) of the suspensions were made using a pH-ORP meter (HANNA HI-4222, St. Louis, MO, USA). An ORP electrode cell combination consisting of a platinum working electrode and a saturated Ag/AgCl reference electrode was used.

2.4. Experimental Design

Two sources of water that provide chloride, sea water (20 g/L Cl\(^-\)) and reject brine (39 g/L Cl\(^-\)), were evaluated. The copper extraction was determined every 4 h until reaching a total time of 48 h. The total mass of calchocite in all tests was always 200 mg. The tests were repeated adding MnO\(_2\) to the reactor with the chalcocite pulp and measuring every 4 h until completing 48 h. Two concentrations of MnO\(_2\) were used, 0.25:1 and 0.5:1, with respect to the total mass of chalcocite. Finally, to evaluate the effect of H\(_2\)SO\(_4\) on all the tests carried out, three concentrations were used, 0.1, 0.5 and 1 M.

3. Results

3.1. Effect of Chloride Concentration on Copper Extraction

The effect of chloride ions in leaching processes has proven beneficial for the treatment of sulphide minerals [14,25,26]. On the other hand, recent studies have shown the benefits of oxidizing primary sulphides incorporating MnO\(_2\) into the system [22–24]. Figure 2 for chalcocite shows better results in chloride-concentrated water such as the brine reject from reverse osmosis plants (ca. 39 g/L of Cl\(^-\)) than in seawater (20 g/L of Cl\(^-\)) both in the presence of MnO\(_2\) in a concentration 0.25:1 w/w with respect to the chalcocite mass. Using reject brine, extractions over 50% Cu are obtained in 8 h and 67.5% Cu in 48 h, apart from the fact that other ions present such as calcium, magnesium and carbonate do not adversely affect the Cu\(_2\)S dissolution. Figure 2 shows that 80% or more of the copper extraction occurs quickly within the first 8 h.

![Figure 2. Cu extraction from chalcocite in two water sources and the addition of MnO\(_2\) in a ratio 0.25:1 w/w with respect to the chalcocite mass.](image-url)
3.2. Effect of MnO₂ Concentration on Copper Extraction

Figure 3 shows the effect of MnO₂ concentration on chalcocite dissolution in reject brine. As the MnO₂ concentration increases, the dissolution of chalcocite increases. This is consistent with the results presented by Devi et al. [22] and Havlik et al. [24], which show that MnO₂ increases the dissolution of even a more refractory sulfide such as chalcopyrite when working at very high concentrations of MnO₂ and HCl. Figure 3 shows that the kinetics of chalcocite dissolution is faster in the presence of MnO₂ but it is more so when the concentration of MnO₂ is higher (ratio 0.5:1 w/w or more with respect to the chalcocite mass). In the latter case, a greater dissolution is obtained at very short times (less than 10 h) which slowly stabilizes at higher dissolution values at longer times.

![Figure 3. Kinetics of chalcocite dissolution in reject brine. Effect of MnO₂ concentration.](image)

Table 4 compares the results of chalcocite dissolution in chlorinated media (sea water and reject brine) in two cases: Without the addition of an oxidizing agent and with the incorporation of a small amount of MnO₂ under the same operating conditions. The table shows that the higher the chloride concentration, the greater the leaching performance, and that the presence of MnO₂ increases the dissolution in all cases. Dissolution data without MnO₂ in Table 4 are from Toro et al. [14]. Note that for reject brine the data is very similar to that shown in Figure 3. MnO₂ increases chalcocite dissolution at short times of 4 h by 8% in sea water and by 10% in reject brine. At long times of 48 h the increase is more moderate, 2% in sea water and 2.5% in reject brine. More interesting are the short time data considering that in a field operation the leaching solution is continuously refreshed.

| Experimental Conditions and Results | Toro et al. [14] | This Work |
|------------------------------------|-----------------|-----------|
| Temperature (°C)                   | 25              | 25        |
| Particle size of Cu₂S (um)         | -147 ± 104      | -147 ± 104|
| H₂SO₄ concentration (mol/L)        | 0.5             | 0.5       |
| MnO₂/Cu₂S ratio (w/w)              | -               | 0.25:1    |
| Dissolution in sea water after 4 h (%) | 32.8         | 35.6       |
| Dissolution in reject brine after 4 h (%) | 36            | 40        |
| Dissolution in sea water after 48 h (%) | 63.4          | 64.7       |
| Dissolution in reject brine after 48 h (%) | 64.6          | 66.2       |

Figure 4 shows the potential for the tests performed in Figure 3. The potential varies between 550 and 630 mV in agreement with Senanayake [33], which indicates that for the dissolution of chalcocite in a chloride-iron solution at 25 °C it is necessary to reach potential values greater than 500 mV. Figure 4 shows higher potentials and copper extractions at short leaching times. Furthermore, Miki et al. [29] have
stated that upon dissolving 50% of chalcocite, the dissolution kinetics becomes slower because the mineral changes phase to covellite, which implies the need to reach potentials greater than 600 mV to dissolve it.

### 3.3. Effect of Sulfuric Acid Concentration on Copper Extraction

The effect of the sulfuric acid concentration on the chalcocite dissolution in reject brine from desalination is presented in Figure 5. In general, the H$_2$SO$_4$ concentration does not have a significant effect on the Cu$_2$S dissolution in a chlorinated medium; this is in agreement with Cheng and Lawson [25] who determined that only a minimal amount of sulfuric acid is needed. An increase in the concentration of sulfuric acid has no significant effect on the dissolution of chalcocite in the presence of MnO$_2$, whatever its concentration.

The results of this study confirm the findings of Toro et al. [14], Dutrizac [34], Cheng and Lawson [25] and Senanayake [26] regarding that high concentrations of chloride favor dissolution kinetics of chalcocite, and of other copper sulfides such as chalcopyrite and covellite [27–29], other variables or conditions are not as relevant. At the same time, the results of this study suggest that the addition of MnO$_2$ enhances the dissolution of chalcocite regardless of the concentration of chlorides.
Figure 5. Effect of sulfuric acid concentration on copper extraction using reject brine, (a) without MnO₂, (b) MnO₂/Cu₂S 0.25:1 w/w and (c) MnO₂/Cu₂S 0.5:1 w/w.

4. Conclusions

Rejection brine from desalination plants and manganese nodules improve the dissolution kinetics of chalcocite, especially at short times. The main conclusions of this work are:

1. Reject brine shows higher Cu extracts from chalcocite compared to sea water due to its higher chloride concentrations.
2. Low concentrations of MnO₂ significantly improve the dissolution kinetics of chalcocite in a short time, which is important for continuous leaching operations.
3. High concentration of H₂SO₄ is not important when working at high concentrations of chloride and MnO₂.
4. Best results are obtained in reject brine, with MnO₂/Cu₂S 0.5:1 w/w and 0.5 mol/L H₂SO₄, that is, 71% Cu extractions in 48 h.
5. Results are expected to contribute to a sustainable process of dissolution of chalcocite.

In future works, these expressions should be replicated but reusing black copper minerals, in order to reuse MnO₂ from waste, and thus justify the additional cost of adding an oxidizing agent. Finally, these new results should be compared with those presented in this study.

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