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HYDROMETALLURGICAL PROCESSING OF EGYPTIAN MAGNESITE ORE

Egyptian magnesite ore (43.32% MgO) from the Eastern Desert was leached with aqueous hydrochloric acid. The following optimum conditions of leaching were determined: ore particle size -0.5 mm, temperature 60 °C, leaching time 15 min, HCl to MgO molar ratio 1.06, and liquid/solid ratio of 2.5:1 ml/g. These conditions lead to recovery of about 99.1% MgO. Resultant liquor was first treated with calculated amount of sulphuric acid in order to precipitate calcium ions and then was neutralized to remove ferric ions. Optimum conditions of this purification procedure were determined. Purified magnesium chloride solution was evaporated and magnesium chloride crystals of 99.5% purity were obtained. Block flowsheet of this process was proposed.

INTRODUCTION

Many processes have been adapted to produce magnesium chloride from magnesites by leaching with hydrochloric acid. Pertl (1973) treated crude magnesite calcined at 800 °C of 0.1–1.0 mm particle size with water followed by passage of HCl gas through this suspension at about 100 °C to produce magnesium chloride. Horak et.al. (1974) leached calcined ore containing more than 8.6% MgO with HCl (16–25%). Harris et. al. (1988) first ground magnesite to 60% less than 6 mesh particle size, calcined it at 700 °C for 2 hours and then treated with 35.5% HCl solution in a stirred tank. On the other hand, Jeny et.al. (1976) dissolved magnesium minerals without calcination countercurrently with 18–21% HCl solutions. Sen et.al. (1984) leached Jhiroli magnesite ore (MgO 45%) of 100% less than 60 mesh particle size at 100 °C for 45 minutes. Mejdal et.al. (1990) leached magnesite with 30% HCl at relatively low temperature of 85–90 °C. Ya et.al. (1990) prepared magnesium chloride from boromagnesite using a salting out process.

In this study a sample of Egyptian magnesite was leached with aqueous hydrochloric acid, treated with calculated amount of sulphuric acid, neutralized and then filtered. Resultant filtrate was evaporated to obtain hydrated magnesium chloride. The block flowsheet of the process is proposed.

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EXPERIMENTAL.

Raw material and apparatus

A sample of magnesite from the Eastern Desert (Egypt) and 32% pure hydrochloric acid were used. The reaction was performed in a 500 ml round bottom thermostated flask. Reaction mixture was agitated at a rate of 500 r.p.m. and then filtered under vacuum in a Buchner-type filter using polypropylene filter cloth of 200 mesh aperture size.

Procedure

Calculated amount of magnesite was added gradually into agitated hydrochloric acid solution in the reaction flask. After the desired reaction time, slurry was treated with calculated amount of sulphuric acid to precipitate calcium, neutralized to pH 7.0 –7.5, and filtered from silica, calcium sulphate, ferric oxide and other impurities. The filtrate was analysed for MgO content and then evaporated. Resultant crystals were analysed for total MgO content. The residue was washed two times with warm water. Filtration and washing times were recorded to calculate the filtration rate. At applied pressure difference 600 mm Hg the rate of filtration (in tonnes of MgO produced per square meter per day) is given by the equation:

$$F = W \cdot 10^{-6} \cdot 86400 / A \cdot 10^{-4} \cdot 0.8T$$

where:

F – filtration rate, tonne MgO/m²·day,

W - weight of MgO in filtered solutions,

A – filter area, cm²,

T - filtration and washing times in seconds,

0.8 - ratio of effective filter area to total filter area.

Characterisation of magnesite ore

Chemical, X-ray diffraction, thermal and sieve analyses of the magnesite ore were performed and the results were published elsewhere, Abdel-Aal et.al. (1994). The sample contained 43.32% of MgO and low contents of CaO, SiO₂ Fe₂O₃ and Al₂O₃. X-ray diffraction of ore indicates that the major mineral is magnesite (MgCO₃) while dolomite (CaCO₃·MgCO₃) and antigorite (3MgO·2SiO₂·2H₂O) are minor minerals. Thermal analysis of the ore shows two endothermic reactions, one at 550 °C due to the decomposition of magnesite:

$$MgCO_3 \xrightarrow{T} MgO + CO_2$$

and the second at 850 °C due to the decomposition of dolomite:

$$CaCO_3 \cdot MgCO_3 \xrightarrow{T} CaO \cdot MgO + 2CO_2$$
.

Particle size distribution of the crushed ore as received reveals that the ore contains little (12.4%) of fine fraction (-0.5 mm). The MgO content in different fractions are roughly similar.

RESULTS AND DISCUSSION

Leaching

Leaching is based on the decomposition of the magnesite with hydrochloric acid according to the following equation:

$$MgCO_3 + 2HCl \longrightarrow MgCl_2 + CO_2 + H_2O$$
.

Moreover, hydrochloric acid reacts with some impurities e.g.:

$$CaCO_3 + 2HCI \longrightarrow CaCl_2 + CO_2 + H_2O$$
,
 $Fe_2O_3 + 6HCI \longrightarrow 2FeCl_3 + 3 H_2O$.

The main factors affecting the leaching are: particle size, reaction temperature, reaction time, liquid/solid ratio and stoichiometric ratio of hydrochloric acid to MgO. These factors were systematically studied.

Effect of particle size

Decrease in the particle size leads to the increase in surface area, which promotes the reactivity. On the other hand, size reduction is a costly operation and should be

Table 1. Effect of particle size of tested magnesite on the MgO recovery

Particle size, % (-0.5 mm fraction*)	MgO recovery %
0	79.10
25	83.90
50	86.60
75	88.40
100	90.65
100**	91.31

^{*} The remaining was -4.0 + 0.5 mm fraction.

avoided, if possible. A series of experiments was carried out using different proportions of ore of different particle sizes from -0.5 mm and -4.0+0.5 mm. Additionally,

^{** 100%} of -0.335 mm fraction.

relatively fine ore (-0.335 mm) was tested. The experiments were conducted at 50 °C for 30 minutes at liquid/solid ratio 4:1 ml/g with about 18% hydrochloric acid and at stoichiometric ratio HCl:MgO = 1. Reaction products were analysed and the MgO content was referred to the original MgO content in the ore in order to calculate the MgO recovery. Results given in Table 1 revealed that the optimum particle size of the magnesite ore for leaching is -0.5 mm (-35 mesh) where 90.65% MgO recovery was achieved. Further grinding of the ore to -0.335 mm particle size gives insignificant increase in MgO recovery.

Effect of reaction temperature

Experiments were performed at temperature ranging from 30 to 80°C. At lower temperatures, where the viscosity of liquid phase is higher which leads to a decrease in the mobility of reacting ions and consequently lowers the reaction rate. On the other hand, higher temperatures result in excessive foam formation together with increasing corrosivity of hydrochloric acid. The leaching was examined under the following conditions:

Particle size - 100% -0.5 mm, Reaction time - 30 min, Liquid/solid ratio - 4:1 ml/g, Stoichiometric ratio HCl/MgO - 1.0, Acid concentration - 18%.

Results are given in Fig. 1 which reveal that the MgO recovery increase with increasing temperature. The optimum temperature is 60 °C which gives 94.18% MgO recovery. Further increase of the temperature gave no increase in MgO recovery.

Effect of reaction time

To study the effect of reaction time on the MgO recovery, experiments were carried out under the above conditions for different periods (5–60 min). Results are given in Fig. 2 which reveal that the reaction of magnesite with hydrochloric acid is a spontaneous reaction of a high rate. Practically, the recovery of MgO equal to 82.23% was obtained after 5 minutes. It is obvious that recovery of MgO increases with the increasing reaction time. The optimum reaction time is 15 minutes where 93.38% of MgO recovery is achieved. Increase of reaction time over 15 minutes gives no appreciable increase in MgO recovery.

Effect of liquid/solid ratio

To study the effect of liquid/solid ratio on MgO recovery and MgCl₂ concentration, experiments were carried out at liquid/solid ratio changing from 6.0:1 to 2.5:1 ml/g. Results are presented in Fig. 3 which shows that higher liquid/solid ratio leads to lower magnesium chloride concentration (14.18%) which requires more energy to evaporate. On the other hand, at lower liquid/solid ratio (2.1:1), the filtrate becomes supersaturated which results in crystallization and losses of MgCl₂ during filtration.

The optimum liquid/solid ratio is 2.5:1 ml/g which gives 94.66% MgO recovery and 32.98% of MgCl₂ in the filtrate.

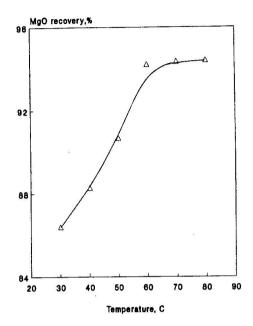
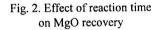


Fig. 1. Effect of temperature on MgO recovery



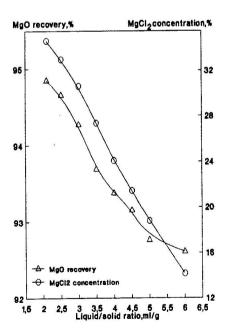


Fig. 3. Effect of liquid/solid ratio on MgO recovery and magnesium chloride concentration

Effect of HCl/MgO stoichiometric molar ratio

The effect of molar ratio of HCl to MgO on MgO recovery and MgCl₂ concentration was studied by changing this ratio from 1.00 to 1.10 under the optimum experimental conditions. Results are given in Fig. 4. The increase in HCl/MgO molar ratio leads to increase of MgO recovery which at ratio 1.06 reaches 99.1%. Further increase in this molar ratio produces an increase of free HCl content in magnesium chloride solution and does not produce an increase in MgO recovery.

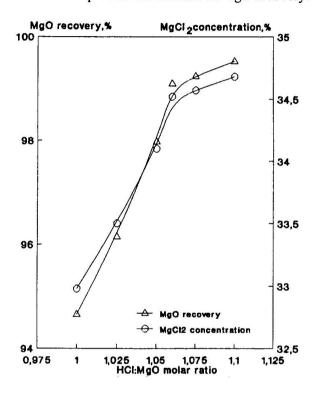


Fig. 4. Effect of HCl/MgO mole ratio on MgO recovery and MgCl₂ concentration

Optimum of the leaching conditions

The optimum conditions and results of decomposition of the magnesite ore with HCl are summarised in Table 2. The maximum MgO recovery achieved is 99.10%.

Purification

Calcium ions, the major dissolved impurity are precipitated as sulphates using sulphuric acid, Jeny (1978) and Grill et. al. (1980). The effect of sulphuric acid to calcium oxide molar ratio on calcium removal efficiency was studied. Other dissolved impurities like iron, aluminium and manganese are removed by precipitation at pH

Conditions:	
Particle size	100%-0.5 mm
Temperature,°C	60
Reaction time, min	15
Liquid/solid ratio, ml/g	2.5:1
HCl acid stoichiometry	1.06
HCl concentration,%	18
Results:	
MgO recovery,%	99.10
MgCl ₂ cocentration,%	34.52

Table 2. Optimum conditions and results of magnesite leaching

7.0-7.5 for 10 minutes at 60 °C by addition of various additives. These were chemically pure magnesium oxide (pH of its slurry was 10.4), magnesite (pH 9.2), magnesite calcined at 550 °C (pH 10.2), magnesite calcined at 850 °C (pH 10.4) and freshly prepared magnesium hydroxide (pH 10.5). The main reactions in this stage are:

$$\begin{aligned} H_2SO_4 + CaCl_2 &\longrightarrow CaSO_4 + 2HCl, \\ MgO + 2HCl &\longrightarrow MgCl_2 + H_2O, \\ MgO + H_2SO_4 &\longrightarrow MgSO_4 + H_2O, \\ \\ 3MgO + 2FeCl_3 + 3H_2O &\longrightarrow 3MgCl_2 + 2Fe(OH)_3, \\ \\ 3MgO + 2AlCl_3 + 3H_2O &\longrightarrow 3MgCl_2 + 2Al(OH)_3. \end{aligned}$$

REMOVAL OF CALCIUM

Effect of H2SO4/CaO mole ratio

Experiments were carried out to study the effect of H₂SO₄/CaO mole ratio on the efficiency of calcium removal under the following conditions:

Temperature: ambient, time: 10 min, acid concentration: 98%.

The results (Table 3) reveal that increasing H_2SO_4 :CaO mole ratio leads to increasing removal efficiency of calcium ions up to 93.3% at 1.1 mole ratio. Increasing H_2SO_4 /CaO mole ratio higher than the optimum value leads to increasing concentrations of free sulphuric acid.

H ₂ SO ₄ ·CaO	·CaO Efficiency of	
mole ratio	CaO removal, %	
1.000	79.30	
1.025	84.83	
1.050	88.95	
1.075	91.73	
1.100	93.30	
1 125	93 63	

Table 3. Effect of H₂SO₄:CaO mole ratio on efficiency of CaO removal

Removal of iron, aluminium and manganese

Iron, aluminium and manganese are removed by adding one of the following materials: pure magnesium oxide, calcined magnesite at 550 °C, calcined magnesite at 850 °C, and freshly prepared magnesium hydroxide securing the final pH from 7.0–8.2. The added amount is 4% of the amount of leached magnesite ore. The total efficiency of purification is 97.4%.

Filtration

The filtration rate of MgCl₂ solution is 2.18 tonne MgO/m² per day under the following conditions: pressure difference – 600 mm Hg, cake thickness– 8 mm, type of filter cloth – polypropylene, slurry temperature – 60 °C.

Table 4. Chemical	analysis of	magnesium	cmorae	product

Constituent	%
MgO	19.74
CI ⁻	34.92
CaO	0.09
SO ₃	0.15
MnO_2	0.008
Fe ₂ O ₃	0.001
Al ₂ O ₃	0.001

Quality of product

The produced magnesium chloride was evaporated till the specific gravity of the solution reached 1.45 g/ml. The solution was cooled and magnesium chloride hexahydrate was crystallized. The resultant MgCl₂·6H₂O was analysed (Table 4). The purity of magnesium chloride crystals was about 99.5% with total dissolved impurities content 0.25%.

CONCLUSIONS

Leaching of Egyptian magnesite ore from Eastern Desert was examined. Recovery of 99.1% MgO was achieved under the following leaching conditions: particle size 100% -0.5 mm, temperature 60 °C, reaction time 15 minutes, liquid/solid ratio 2.5:1

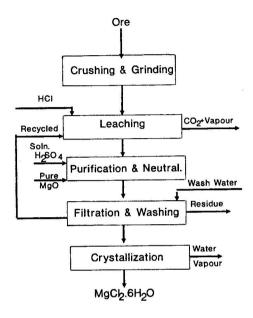


Fig. 5. Block flowsheet for production of magnesium chloride from Egyptiam magnesite ore

ml/g and hydrochloric acid to magnesium oxide molar ratio 1.06. The resultant slurry was treated with H₂SO₄ to precipitate calcium ions then neutralized to remove iron, manganese and aluminium impurities. The slurry was then filtered at the rate of 2.18 tonne MgO/m²·day at 600 mm Hg pressure difference. The purity of produced magnesium chloride crystals reaches 99.5%. The proposed process flowsheet is given in Fig. 5.

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Ługowano za pomocą 32% roztworu kwasu solnego rudę magnezytową z Pustyni Wschodniej zawierającą 43% MgO. Określono, że optymalne warunki procesu to: wielkość ziarn –0,5 mm, temperatura 60 °C, czas ługowania 15 minut, stosunek molowy HCl do MgO 1,06 oraz stosunek fazy ciekłej do stałej jak 2,5:1 ml/g. Warunki te pozwoliły na odzyskanie około 99,1% MgO. Otrzymany w wyniku ługowania roztwór traktowano obliczoną ilością kwasu siarkowego w celu wytrącenia jonów wapnia a następnie roztwór neutralizowano w celu usunięcia jonów żelaza. Określono także optymalne warunki tej operacji. Oczyszczony roztwór chlorku magnezu poddano odparowaniu i otrzymano krystaliczny MgCl₂ o czystości 99,5%. W pracy zaproponowano blokowy schemat przerobu badanych rud.