

# Improvement of the EAF's energetic parameters using capacitive-inductive filters

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*Abstract:* - In this paper is analyzing the current operating conditions of one electric arc furnace (EAF) in order to evaluate the best option to solve the energy consumption problem.

Experimental results show that EAFs represent a substantial source of electric disturbances, such as voltage fluctuations, flicker, harmonics, and unbalance between phases. Improvement of the energetic performances of an EAF imposes a careful technical and economical analysis.

The possible compensation solutions include passive filter, SVC (static var compensator) and STATCOM (static synchronous compensator).

Finally, the guideline for compensator selection and performance estimation is obtained.

*Key-Words:* - Electric Arc Furnace, Flicker, Power Factor, Harmonic Analysis, Reactive Compensator, Improvement.

## 1 Introduction

A great percent of the world production is provided by large capacity electric arc furnaces (EAF). EAF are placed among the biggest polluters of air, soil, water and electric supply grids. Also the energy consumption is significant [1].

The ability to precisely control the temperature and chemistry of the batch make EAFs an ideal choice for producing high-grade steel for the recycling of scrap. Of the steel made today 36% is produced by the EAF route and this share will increase to 50% by 2030 [2], [3].

In contrast to other types of loads which are usually operated by voltage steps, EAFs produce random flicker which cannot be easily calculated with standard curves and methods [4].

The electrical design of a modern arc furnace installation involves a study encompassing the complete power circuit from the utility company's generators to the arcs in the furnace. The study also entails the selection of suitable circuit components as required to put the desired power into the furnace at optimum conditions and to restrict cyclic voltage changes (flicker) to acceptable limits at some designated point in the power system and correct power factor as required. Occasionally, someone will suggest the use of a higher impedance step-down transformer between

the utility system and the high-voltage furnace bus to reduce the surges on the utility system. This reasoning is completely erroneous, as the furnace builder would simply select a higher no-load furnace voltage to maintain the desired arc characteristics. This, if anything, would produce slightly larger swings between no-load and short circuit [5] - [8].

## 2 Solutions for flicker mitigation

From the power flow point of view, the basic principle of the flicker mitigation solution can be simply explained as shown in the Fig. 1. The power consumed by EAF can be regarded as a constant power ( $P_C$ ,  $Q_C$ ) plus a fluctuating power ( $\Delta P_f$ ,  $\Delta Q_f$ ). Generally,  $P_C$  affects the angle stability,  $Q_C$  affects the bus voltage profile, voltage stability, and load power factor.  $\Delta P_f$  is mainly related to the fluctuations of bus voltage angle while  $\Delta Q_f$  is mainly related to the fluctuations of bus voltage magnitude [9]. An obvious total solution to solve the EAF power quality issue is to compensate  $Q_C$  by amounts  $\Delta P_f$  and  $\Delta Q_f$  so that the source only supplies constant  $P_C$  at unity power factor and so that the source bus voltage has a constant magnitude and angle. However, a total solution is not necessary and a cost-effective alternative is to provide compensation by supplying  $\Delta P_f$  and  $\Delta Q_f$  only in the

troublesome frequency range, that is, the EAF flicker frequency range of 1Hz ~ 20 Hz [9].

How to economically and efficiently mitigate EAF flicker is consistently a tough issue for utility and industry professionals. The basic methodology for flicker mitigation can be categorized into three types [10]:

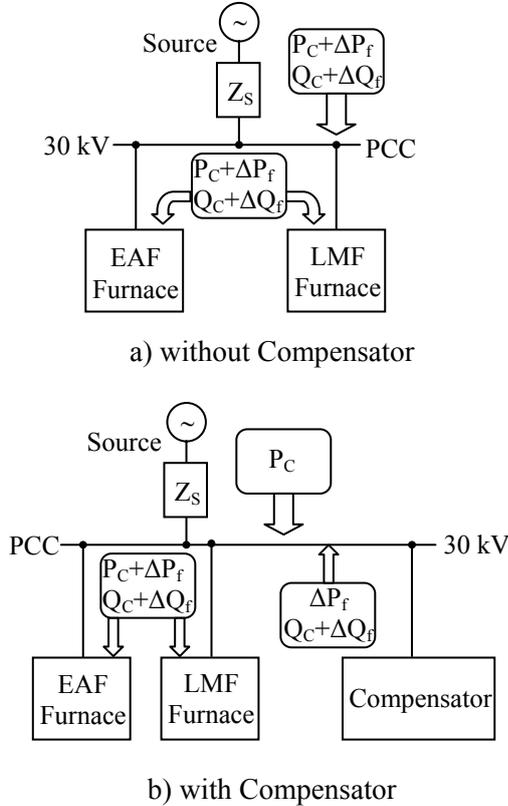


Fig. 1. Arc Furnace Compensation Block Diagram

- passive filters, which can be either series or shunt;
- series active compensator, such as the series impedance regulation;
- shunt active compensator, such as SVC (static var compensator) and STATCOM (static synchronous compensator).

Though passive filters are simple, reliable, low-cost and highly efficient, it is difficult to design for a stiff system, time-consuming for tuning, easy to induce resonance, and susceptible to system impedance variations [10].

Although the increasing series reactance can mitigate the flicker, it results in the voltage reduction and therefore decreases EAF productivity. Moreover, it is also expensive and laborious to control upstream transformer reactance or series reactor in the firmer deregulation power system [10].

SVC can improve system power quality, and also increase EAF productivity and provide additional economic benefits. However, it cannot react to the

fast-varying flicker very well with the inherent limit of relatively low bandwidth and hence its dynamic performance for flicker mitigation is limited. The STATCOM is based on high frequency switching voltage-source converter (VSC). While SVC performs as a controlled reactive admittance, STATCOM functions as a synchronous voltage source.

### 3 System description and experimental investigation

A one-line diagram of the studied system is shown in Fig. 2. The 75 MVA arc furnace is supplied from a local feeder with a 30 kV bus voltage. The local feeder is in turn supplied from a 220 kV transmission line through a 160 MVA, 220/30 kV transformer. The compensator is planned to be installed on the EAF 30 kV bus (PCC) as this is the most effective location, primarily because of voltage magnitude considerations.

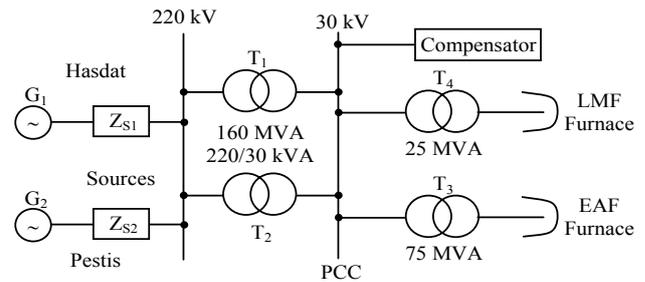


Fig. 2 One-line diagram of the system

Typically, the EAF's major processes are melting (early melt) and the refining (late melt). During these two processes, the EAF's electrodes are short-circuited to scrap metal and consume time-varying power, which is mainly dependent on arc lengths.

In table I are given the characteristics of the sources and of the T<sub>1</sub>, T<sub>2</sub>, T<sub>3</sub> and T<sub>4</sub> transformers [11].

This research theme was required by the beneficiary because the costs with the reactive energy are very high. In addition, if the reactive energy consumption is compensated, the power factor increases and the furnace has available a higher power for melting, the average heat time being reduced.

In table II are presented the additional costs with the reactive energy in 2008.

Taking into account the total values paid for the reactive energy in one year (827700.29 Euro at a power factor of 92%) and the system operator intends to increase the value of the neutral power factor to 94% (for which would be paid 944855.72

Euro) is imposing the urgent adaption of a solution and its practical implementation.

Table I

Concept	Unit		
Source		Pestis	Hasdat
Reference Voltage	kV	220	220
Frequency	Hz	50	50
Reactance	PU	0.02088	0.01871
Resistance	PU	0.00182	0.00163
Impedance	PU	0.02096	0.01879
X/R		11.5	11.5
Step Down Transformer		T <sub>1</sub>	T <sub>2</sub>
Impedance	%Z	16.38	16.38
Capacity	MVA	160	160
Reference Voltage	kV	239.2	239.2
HV Tap	kV	239.2	239.2
LV Tap	kV	33	33
X/R		19.91	19.91
Impedance	Ohm	58.576	58.576
Source+Furnace Transformer		T <sub>3</sub>	T <sub>4</sub>
Secondary Voltage	V	750	690.7
Capacity	MVA	75	25
X/R		17.358	21.167
Impedance	PU	0.0365	0.0511
HV Connection		Delta	Delta/Wye
LV Connection		Delta/Wye	Delta/Delta

Table II

2008 Month	Cumulated Active Energy [MWh]	Cumulated Reactive Energy [MVarh]	Power factor	Excess of Reactive Energy [MVarh]				Billing [Euro]	
				K <sub>p</sub> = 92%	K <sub>p</sub> = 94%	K <sub>p</sub> = 92%	K <sub>p</sub> = 94%		
1	24949.91	22194.96	74.72	11.533.34	13139.35	76466.37	86865.71		
2	47714.63	40754.13	77.51	8861.44	10296.68	58584	68072.53		
3	71788.99	63712.55	72.37	12702.78	14220.59	83979.51	94013.91		
4	97815.20	97613.33	73.65	12813.66	14454.52	84712.54	95560.49		
5	124438.45	110151.20	76.32	11196.41	12874.91	74020.72	85117.52		
6	142990.92	125643.81	76.76	7589.28	8758.95	50173.62	57906.45		
7	164626.42	144399.16	75.56	9538.66	10902.71	63061.20	72079.06		
8	183980.63	160763.03	76.36	8119	9339.22	53675.65	61742.66		
9	204818.46	178251.05	76.60	8611	9924.89	56929.23	65614.61		
10	232389.63	202178.96	75.52	12182.64	13920.91	80540.80	92032.70		
11	254595.56	221729.83	75.05	10091.31	11491.32	66714.81	75970.43		
12	277698.01	243625.15	74.11	11925.65	13595.24	78841.84	89879.65		
Total				125198.17	142919.29	827700.29	944855.72		

In table III are presented the main production indicators obtained in 2008.

Table III

EAF Concept	Unit	Month- 2008												Average
		1	2	3	4	5	6	7	8	9	10	11	12	
Production (Finished Product)	ton	37,020	32,080	33,719	38,125	39,614	25,235	28,736	25,641	28,213	38,535	30,686	28,436	32,170
Heats		341.0	299	314	360	369	247	281	251	266	361	287	269	304
Heat Production (Finished)	ton/Heat	108.6	107.3	107.4	105.9	107.4	102.2	102.3	102.2	106.1	106.7	106.9	105.7	105.7
Scrap Charged	ton/Heat	129.94	125.71	125.26	125.40	125.81	126.03	127.23	125.94	125.84	124.89	124.3	124.5	125.9
Yield (Finished Product)	%	0.835	0.853	0.857	0.845	0.853	0.811	0.804	0.811	0.843	0.855	0.860	0.849	0.840
Tap-to-Tap Delays	min/Heat	120.1	127.2	125.2	110.7	111.0	124.9	128.1	129.1	120.2	115.4	126.6	122.4	121.7
Melting Power On Time	min/Heat	60.1	63.7	66.7	54.2	53.6	64.1	66.2	68.6	58.8	52.8	63.6	61.6	61.2
Energy Consumption	Wh/Ton	578.3	599.4	596.5	591.0	577.6	635.9	642.1	642.6	620.7	614.1	609.0	607.1	609.5
Electrode Consumption	Kg/Ton	2.64	2.68	2.59	2.69	2.51	2.18	2.78	2.93	3.41	2.39	2.97	2.41	2.68
Gas Consumption	Nm3/Ton	9.6	14.6	14.8	12.2	10.5	15.3	14.1	14.7	13.6	9.2	10.9	14.1	12.8
Oxygen Consumption	Nm3/Ton	16.8	17.0	34.6	34.3	33.7	29.1	41.8	33.5	37.6	26.3	25.7	23.1	29.5
Carbon EAF Charge	kg/Ton	5.1	10.1	11.0	13.5	12.7	9.8	10.7	11.8	8.2	10.8	9.7	13.3	10.6

To compensate the multiple EAF in Operation it is applied the Jenkin Method as follows:

$$S_{eq} = S_{Largest} \times \sqrt{M_1 + c\sqrt{M_2}}, \quad (1)$$

$$M_i = K_i \cdot \left[ \sum_{EAF=1}^N \left( \frac{S_{EAF}^{(2xi)}}{S_{Largest}} \right) \right], \quad (2)$$

$$b = \frac{M_3^2}{M_2^2}, \quad (3)$$

where  $S_{eq}$  – is equivalent size of EAF, [MVA];  
 $S_{Largest}$  - size of the largest EAF, [MVA];  
 $S_{EAF}$  - size of each EAF, [MVA];  
 $M_{1,2,3}$  - statistical moments;  
 $K_{1,2,3}$  - weighing factors;  
 $N$  - total EAF's in the system;  
 $c$  - stretch factor (from chart of Guide Number);  
 $b$  - Guide Number.

For our configuration  $S_{Largest} = 75$  MVA and  $S_{EAF LMF} = 25$ MVA, results

$$S_{eq} = 75.84 MVA. \quad (4)$$

In order to propose a solution for improving the operation and a compensation solution, there were made experimental measurements, a part of the results being presented in the next figures [11].

In fig. 3 is presented the variation form of the line voltages on the bar of 30 kV. Against the nominal value takes place a variation comprised within  $\pm 10\%$ . The more pronounced variations can be observed in the melting phase (early melt). There are significant differences between voltages caused by the working mode of the electrodes and their operation regime.

The current's variation form in the EAF transformer's secondary and the arc's stability are given in fig. 4. One can notice that the great current variations determine a pronounced instability of the electric arc. The variation of the active and reactive power depending on the current in secondary during the melting itself (early melt) is presented in fig. 5 and 6. The average active power is around 62 MW and the average reactive power 47 MVar.

The dependency of the power factor on the current in secondary for the two melting phases (early melt and late melt) is given in fig. 7 and 8. In the first case (fig.7) results an average value of 78% and in the second case of 84%.

The measurements were made during more heats, and the presented results were selected from the most suggestive ones [11].

In fig. 5, 6, 7 (early melt) the current's variation field is large (40÷86 kA) and in fig. 8 (late melt) this field is restraining (56÷60 kA). Due to the smaller current variations and increased stability in operation (fig. 4), takes place a reduction of the reactive energy consumption (from an average of 47 MVar to 43 MVar), an increase of the power factor

(by 6% the average value) and an increase of the average active power (from 62 MW to 69 MW).

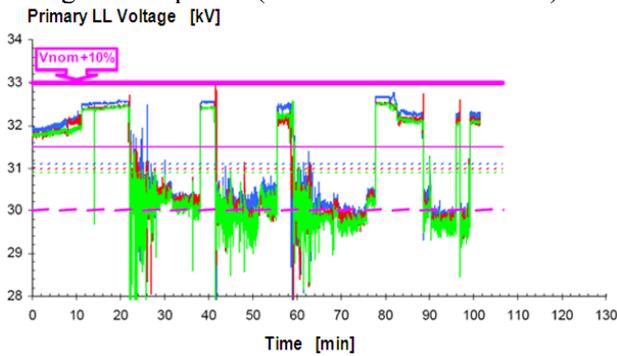


Fig. 3 Line voltages in the EAF transformer's primary

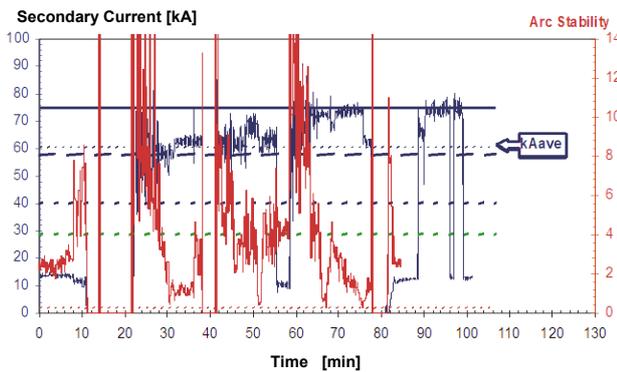


Fig. 4 Current in the secondary and electric arc's stability

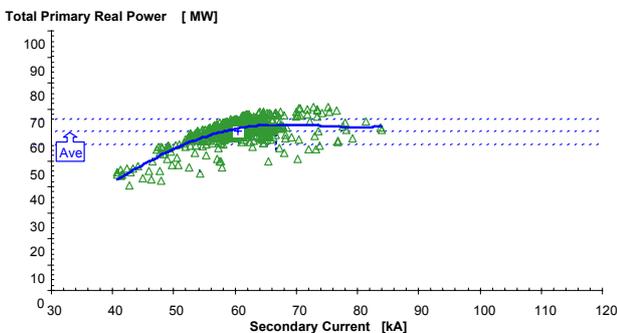


Fig. 5 The active power variation curve in the melting phase (early melt)

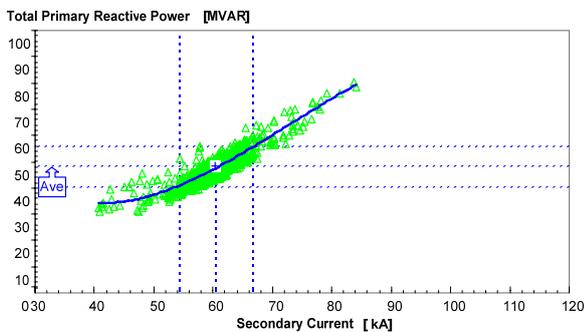


Fig. 6 The reactive power variation curve in the melting phase (early melt)

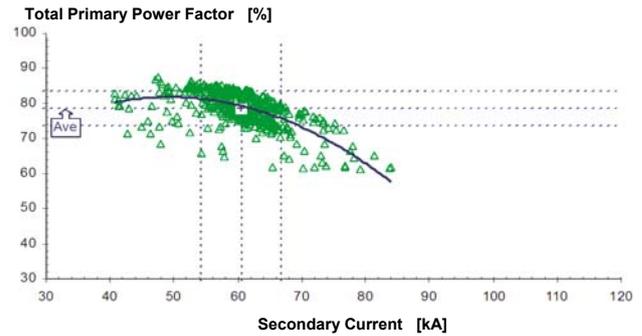


Fig. 7 Power factor in the EAF's primary (early melt)

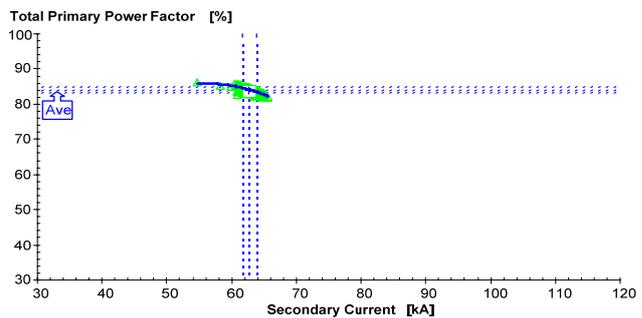


Fig. 8 Power factor in the EAF's primary (late melt)

The scope of study it's to analyze the PF improve, identify operating parameters expected in the future, and evaluate [11]:

- EAF performance from the point of view of electrical energy with recommendations to improve;
- recommendations regarding the power factor problem (MVAR required, bank configuration, number of filters and tuning reactor specification);
- evaluate existing electrode regulator performance and point to improvement opportunities;
- recommend the best operating set-points to run its EAF efficiently;
- identify operative characteristics and system restrictions in different heat stages.

#### 4 Solution for EAF improvement performances

From the collected data analysis we can notice several possibilities for improvement: power factor compensation with passive elements (capacitors and inductive elements) or active compensation with SVC. Further, are presented the simulations results obtained with compensation with passive elements were considered four situations: one without series reactor (case  $R = 0$ ) and three with series reactor having the resistances of 1, 2 and 3 ohm.

Currently, the variation of the absorbed power, the unbalance on phases, the flicker effect and the distortion of the wave forms of the voltages and currents is not imputed to the consumer only by the reactive energy absorbed to a power factor smaller than the neutral one ( $K_p = 92\%$ ).

However, the modern norms will take into account all these aspects, and the operation costs with such unbalances will increase spectacularly. Therefore is necessary the thinking of some systems that should allow the alignment to the new standards.

In fig. 9 is presented a proposal of automatic system for compensation, filtration and balancing.

There were made the following notations:

- $T_1, T_2$  – power transformers (one active and one for spare);
- $T_3$  – EAF's transformer;
- $T_4$  – LMF's transformer;
- TC – current transformers;
- TT – voltage transformers;
- EAF – electric arc furnace for melting;
- LMF - electric arc furnace for treatment and alloying;
- E – electrodes;
- AB – adaption block;
- PLC – programable logic controller for compensation, filtration and balancing;
- FBSH – filtration block for superior harmonics;
- RECB – reactive energy compensation block with reactances of fixed capacity;
- ARECB – additional reactive energy compensation block and load balancing.

The PLC receives the current and voltage information from the current and voltage transformers, for each phase, by means of the adaption block. Based on the implemented management program, is controlled the harmonics filtration, respectively the compensation of the reactive energy and load balancing on phases. The harmonics' filtration block contains coils and capacitors tuned on the frequency corresponding to the harmonics 3, 5, 7, 11, 13 which in practice was found that they have a higher proportion.

Their connection or disconnection is made by static contactors controlled by the PLC.

The reactive energy's compensation block with fixed capacity reactance's (RECB) is connected in circuit during the entire heat and, is calculated in such way that at the average value of the reactive energy on a heat to be achieved a neutral power factor. When the furnace is in stand-by, this block is disconnected.

The additional compensation block of the reactive energy (ARECB) introduces and takes out dynamically, at the command given by the PLC, by static contactors, batteries of capacitors for compensating the reactive energy that exceeds the

average on a heat in such way that the power factor to not decrease under the neutral value. Also within this block there is the load balancing installation with coils and capacitors, which in real time introduces or takes out reactance's from the circuit in such way that the distortion factor of the currents respectively voltages to be under the permitted limits and the values of the phase differences between currents and between voltages to correspond to a symmetric system. In this block have plane important transitory phenomena. The state-of-the-art installations are using frequency converters that create a capacitive or inductive regime depending on the process requirement, instead of the ARECB block.

In table IV are presented the results of the simulations made for compensating the reactive energy by introducing of capacitor batteries with power between 20 MVar and 50 MVar. Is considered the case of compensation up to the power factor of 94%, taking into account the reactive energy consumptions in 2008 [11].

In the first case ( $K_p = 92\%$ ) a capacitor battery having 40 MVar is sufficient for the entire reactive energy consumption and in the second case ( $K_p=94\%$ ) the value is of 45 MVar (table IV).

One can notice that the available active power at a certain current through the electric arc is more smaller as the voltage in secondary is smaller and the reactor's resistance is higher.

The average capacitor requirement at 30kV to fit 94% power factor are 36.1 MVar at early melt and 30.5 MVar at late melt. Recommended size is 45 MVar at 30kV, providing more than 94% power factor compensation. It is recommended to use the complete 45 MVar in one filter tuned on the 3rd harmonic order (about 2.9, depending upon resonance conditions).

The modernization work is in progress and no solution was finally considered.

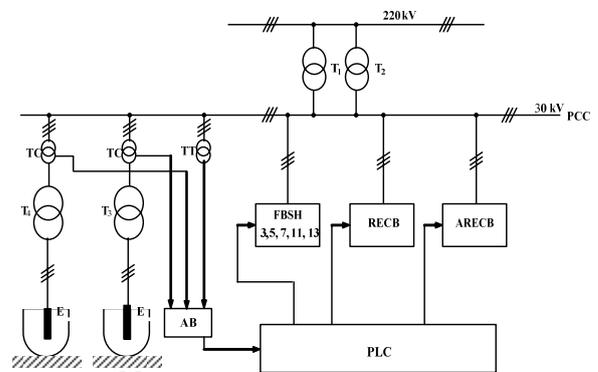


Fig. 9 Block diagram of the balancing and compensation system

Table IV

2008 Month	Kp = 92%										Required Reactive Power [MVar]	Equivalent Capacitor [kVar]	Kp = 94%									
	Required Reactive Power [MVar]	Equivalent Capacitor [kVar]	Missing MVar Compensation (-)										Required Reactive Power [MVar]	Equivalent Capacitor [kVar]	Missing MVar Compensation (-)							
			Size of Capacitor Bank [kVar]												Size of Capacitor Bank [kVar]							
			20000	25000	30000	35000	40000	45000	50000	20000					25000	30000	35000	40000	45000	50000		
1	11,566.34	34,790	-4,917	-3,255	-1,592	70	1,732	3,394	5,057	13,139.35	39,521	-6,490	-4,828	-3,166	-1,503	159	1,821	3,484				
2	8,861.45	29,616	-2,877	-1,381	115	1,611	3,107	4,603	6,099	10,296.69	34,412	-4,312	-2,816	-1,320	176	1,672	3,168	4,664				
3	12,702.78	38,208	-6,054	-4,391	-2,729	-1,067	596	2,258	3,920	14,220.59	42,774	-7,571	-5,909	-4,247	-2,584	-922	740	2,402				
4	12,813.66	39,871	-6,386	-4,779	-3,172	-1,565	42	1,648	3,255	14,454.53	44,977	-8,027	-6,420	-4,813	-3,206	-1,599	8	1,614				
5	11,196.41	33,677	-4,547	-2,885	-1,223	440	2,102	3,764	5,427	12,874.92	38,726	-6,226	-4,563	-2,901	-1,239	424	2,086	3,748				
6	7,589.29	23,615	-1,162	445	2,052	3,659	5,266	6,873	8,480	8,758.96	27,254	-2,331	-724	882	2,489	4,096	5,703	7,310				
7	9,538.67	28,691	-2,889	-1,227	435	2,097	3,760	5,422	7,084	10,902.72	32,794	-4,253	-2,591	-929	733	2,396	4,058	5,720				
8	8,119.01	24,421	-1,470	193	1,855	3,517	5,179	6,842	8,504	9,339.23	28,091	-2,690	-1,028	635	2,297	3,959	5,622	7,284				
9	8,611.14	26,794	-2,184	-577	1,030	2,637	4,244	5,851	7,458	9,924.90	30,882	-3,497	-1,890	-284	1,323	2,930	4,537	6,144				
10	12,182.64	36,644	-5,533	-3,871	-2,209	-546	1,116	2,778	4,440	13,920.91	41,872	-7,272	-5,609	-3,947	-2,285	-622	1,040	2,702				
11	10,091.32	31,400	-3,664	-2,057	-450	1,157	2,764	4,371	5,978	11,491.33	35,756	-5,064	-3,457	-1,850	-243	1,364	2,971	4,578				
12	10,297.52	31,612	-3,789	-2,162	-535	1,092	2,719	4,346	5,973	11,756.74	36,096	-5,249	-3,622	-1,994	-367	1,260	2,887	4,514				

## 5. Conclusions

In this paper, the 100t Electric Arc Furnace (EAF) for steel production is analyzed. The flicker mitigation and reactive power compensation requirements are quantitatively analyzed. The various compensation solutions are compared, such as, passive filter, SVC, and STATCOM.

Supplementary series reactors are widely used in high power AC EAF to allow furnace operation with longer arcs, lower currents and lower electrode consumption. Such reactors can be switched off in the final stages of the melting process, when the electric arc is stable and arc ignition is not of great concern.

A menu for optimum operation of the EAF system should be prepared in terms of EAF transformer and series reactor taps and electrode current settings for different phases of the melting process.

Using the compensation of reactive energy, harmonic filtering and load balancing determines the increase of the power factor more than neutral value (92%), reduction of the harmonic emission in the electric grid, reduction flicker and deforming effect on the grid.

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