Ultimate Boundedness of Droop Controlled Microgrids with Secondary Loops

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Abstract—In this paper we study theoretical properties of inverter-based microgrids controlled via primary and secondary loops. Stability of these microgrids has been the subject of a number of recent studies. Conventional approaches based on standard hierarchical control rely on time-scale separation between primary and secondary control loops to show local stability of equilibria. In this paper we show that (i) frequency regulation can be ensured without assuming time-scale separation and, (ii) ultimate boundedness of the trajectories starting inside a region of the state space can be guaranteed under a condition on the inverters power injection errors. The trajectory ultimate bound can be computed by simple iterations of a nonlinear mapping and provides a certificate of the overall performance of the controlled microgrid.

I. INTRODUCTION

In the last decade, the need to mitigate the environmental impacts of coal-fired electricity generation has stimulated a gradual transition from large centralised energy grids towards small-scale distributed generation (DG) of power [17]. A common operating regime for DG is to form microgrids before being connected to the main energy grid. A microgrid is a small-scale power system consisting of a collection of DG units, loads and local storage, operating together with energy management, control and protection devices and associated software [11], [13].

Control strategies are indispensable to provide stability in microgrids [12]. Recently, hierarchical control for microgrids has been proposed in order to standardise their operation and functionalities [7], [4]. In this hierarchical approach, three main control levels are defined to manage voltage and frequency stability and regulation, and power flow and economic optimisation. In this paper we focus on the primary and secondary control levels, which are the main parts of the automatic control system for the microgrid.

The primary control level deals with the local control loops of the DG sources. Many of these sources generate either variable frequency AC power or DC power, and are interfaced with an AC grid via power electronic DC/AC inverters. For inductive lines, inverters are typically controlled to emulate the droop characteristic of synchronous generators. Conventionally, the frequency-active power (or “ω-P”) droop control [5] is adopted as the decentralised control strategy for the autonomous active power sharing at primary level. Because standard droop control is a purely proportional control strategy, the secondary control level has the task of compensating for frequency steady-state errors induced by the primary control layer. Although the secondary control level is conventionally implemented in a centralised fashion, several recent works have suggested distributed control implementations [15], [3], [14].

Stability and convergence properties of droop-controlled networks of inverters and loads have recently been the focus of the detailed analyses that highlight the dynamic properties of the power system [2], [3], [16]. For example, in [16], the authors present a necessary and sufficient condition for the existence of a unique and locally exponentially stable steady state equilibrium for a droop-controlled network. The paper also proposes a distributed secondary-control scheme to dynamically regulate the network frequency to a nominal value while maintaining proportional power sharing among the inverters, and without assuming time-scale separation between primary and secondary control loops. This is in contrast with more conventional analyses which rely on time-scale separation and do not discuss stability properties beyond local results around equilibrium points [16].

In this paper we analyse ultimate boundedness of the states of an inverter-based purely inductive microgrid with decentralised droop control and secondary control systems. The network of our study is inherently decentralised as no communication between neighbouring droop controllers is needed. Our first contribution is a structured nonlinear model for a microgrid with embedded primary and secondary control levels. By performing a suitable change of coordinates, we show how the stability analysis for the controlled system is decoupled into a linear system stability problem, and that of characterising ultimate boundedness of the trajectories of a perturbed nonlinear subsystem around steady-state solutions. Our second and main contribution is then to establish stability properties of the original nonlinear system by exploiting this model separation. The linear analysis shows that frequency regulation is ensured without the need for time-scale separation. For the perturbed nonlinear subsystem, we show that ultimate boundedness of the trajectories starting inside a region of the state space is guaranteed under a condition on the power injection errors for the inverters. The ultimate bounds for the trajectories can be computed by iterating a well-specified nonlinear map, which provides key certificates for the overall performance of the controlled microgrid.

Notation and Definitions: Let $1_n$ and $0_n$ be the $n$-
dimensional vectors of unit and zero entries. Let $\mathcal{I} = \{1, 2, \ldots, n\}$ and $\mathcal{J} = \{1, 2, \ldots, m\}$ be index sets of inverter buses and edges, respectively. For a matrix $M$, $M_{(i,j)}$, $M_{(i,:)}$, $M_{(:,j)}$, denote its $i$-th row, $j$-th column, rows $i$ to $j$, and $i$-th entry, respectively. Denote by $B \in \mathbb{R}^{n \times m}$ the *incidence* matrix of a directed graph such that $B_{(i,j)} = 1$ if the node $i$ is the source of the edge $j$ and $B_{(i,j)} = -1$ if the node $i$ is the sink node of the edge $j$; all other entries are zero. The *Laplacian* matrix is $L = BYB^T$ where $Y = \text{diag}\{\{a_{ij}\}_{i,j} \in \mathcal{J}\}$, $a_{ij} = y_{ij}E_iE_j$, $y_{ij}$ denoting the pure imaginary $i$-$j$-th line admittance and $E_i$ denoting the bus voltage magnitude. For connected graphs, $\ker B^T = \ker L = 1_n$. The entries of the $m \times 1$ vector function $f = [f(\theta_i - \theta_j)]_{i,j} \in \mathcal{J}$ contain the scalar function $f(\cdot)$ applied to $(\theta_i - \theta_j)$ in the same order as the entries in the matrix $Y$. The symbol $\otimes$ denotes the Kronecker product of matrices. $\mathbb{R}_{+}^{n \times m}$ denotes the set of real $n \times m$ vectors with nonnegative components. $\mathbb{Z}_+$ denotes the set of positive integers. Inequalities and absolute values are taken componentwise. A nonnegative vector function $T : \mathbb{R}_{+}^{n} \rightarrow \mathbb{R}_{+}^{1 \times 0}$ is said to be componentwise nonincreasing (CNI) if whenever $z_1, z_2 \in \mathbb{R}_{+}^{1 \times 0}$ and $z_1 \leq z_2$, then $T(z_1) \leq T(z_2)$.

II. Decentralised Droop Control Model

We start by presenting our model of an inverter-based microgrid under decentralised droop control, and then analyse its structure to reveal important modal characteristics of the underlying linear part of the system. The model is essentially a weighted graph where each node represents a common-voltage point of power injection, and branches represent microgrid node-interconnecting lines [16], [1].

The standard primary droop control at each inverter $i$ in the microgrid is such that the deviation in frequency $\theta_i$ from a nominal rated frequency $\omega^*$ is proportional to the power injection $P_{e,i}$ in the following way:

$$d_i\theta_i = P_i^* - P_{e,i}$$

where $d_i > 0$ is the droop controller coefficient, $P_i^* = P_{ref,i} - P_{e,i}$ is the inverter power injection error between the inverter nominal power injection setpoint $P_{ref,i}$ and the bus load $P_{L,i}$, and $\omega_i = \omega^* + \theta_i$ is the frequency of the voltage signal at the $i$-th inverter. By assuming purely (lossless) inductive lines, the power injection to each bus has the form

$$P_{e,i} = \sum_{j=1}^{n} a_{ij} \sin(\theta_i - \theta_j),$$

with $a_{ij} = y_{ij}E_iE_j$, $y_{ij}$ denoting the pure imaginary $i$-$j$-th line admittance and $E_i$ denoting the bus voltage magnitude. We make the standard *decoupling approximation* [19] where all voltage magnitudes $E_i$ are constant so that the power injection is considered a function of only the phase angles, that is, $P_{e,i} = P_{e,i}(\theta)$.

The droop controller (1) results in a static error in the steady state frequency. In [1], it is shown that as long as the network state trajectories remain in a specified region, then the controller in (1) ensures network synchronisation to the average frequency error

$$\omega_{sync} = \frac{\sum_{i=1}^{n} d_i\dot{\theta}_i}{n} = \frac{\sum_{i=1}^{n} P_i^*}{n}$$

where the last equality follows from the fact that $\sum_{i=1}^{n} P_{e,i} = 0$ for purely inductive lines.

We observe that $\omega_{sync} = 0$ if and only if $\sum_{i=1}^{n} P_i^* = 0$ or equivalently $\sum_{i=1}^{n} P_{ref,i} = \sum_{i=1}^{n} P_{e,i}$, that is, the nominal injections are balanced. As discussed in [6], it is not possible to achieve balanced nominal power injections since they depend on generally unknown and variable load demand. Also, selecting the droop coefficients $d_i$ arbitrary large to make $\omega_{sync}$ small is not realistic. Thus, complementary control action is required to eliminate or at least reduce the frequency error $\omega_{sync}$. For example, by including additional secondary control inputs $p_i$ to each inverter bus as follows:

$$d_i\dot{\theta}_i = P_i^* - \sum_{j=1}^{n} a_{ij} \sin(\theta_i - \theta_j) - p_i,$$

$$k_i p_i = \dot{\theta}_i - \epsilon p_i,$$

for each $i \in \mathcal{I}$ with $k_i, \epsilon > 0$. As shown in [2] and discussed here in Section III-C the parameter $\epsilon$ in [5] can be tuned to reduce the frequency error.

**Assumption 2.1:** In this paper we take all the droop coefficients as well as all the secondary control coefficients to be identical, that is, $d_i = d$ and $k_i = k$ for all $i \in \mathcal{I}$.

The above assumption leads to having a simplified expression for the average frequency error which is

$$\omega_{sync} = \frac{\sum_{i=1}^{n} \dot{\theta}_i}{n}.$$ (6)

Let $\sin(x) = x + f(x)$ where $f(x) = \sin(x) - x$. Then, from the definitions of the incidence matrix $B$ and the Laplacian matrix $L = BYB^T$ introduced in Notation and Definitions above, the system (4), (5) can be expressed as

$$\dot{x} = Ax + Hf + \bar{P}$$ (7)

where $x = [x_0^T, x_p^T]^T$, $x_0 = [\theta_1, \ldots, \theta_n]^T$, $x_p = [p_1, \ldots, p_n]^T$, $f = [f(\theta_i - \theta_j)]_{i,j,\in,J}$ and the matrices

$$A = \begin{bmatrix} \frac{-1}{dL} & \frac{-1}{dL}I_n \\ \frac{1}{dk}L & \frac{-1}{dk}I_n \end{bmatrix}, H = \begin{bmatrix} \frac{1}{dL} BY \\ \frac{1}{dk} P^* \end{bmatrix}, \bar{P} = \begin{bmatrix} \frac{1}{dL} P^* \end{bmatrix}$$ (8)

where $P^* = [P_1 \ldots P_n]^T$ and $\epsilon = 1 + ed$.

Let $(\mu_i, u_i), i \in \mathcal{I}$ be the eigenvalue-eigenVector pairs of the Laplacian matrix $L$ and define the associated eigenvalue and eigenvector matrices as

$$M = \text{diag}\{\mu_1, \ldots, \mu_n\}, \quad U = [u_1 \ldots u_n].$$ (9)

The following properties of the Laplacian eigenstructure will be useful for our later developments.

**Lemma 2.2:** (Properties of the Laplacian eigenstructure)

The Laplacian eigenstructure (9) has the following properties:

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The eigenstructure of the system (7)–(8) can be conveniently represented in terms of the eigenstructure of the Laplacian matrix, as shown in the following theorem.

**Theorem 2.3:** For system (7)–(8), the eigenvalues $\lambda_i$ and eigenvectors $v_i$ of the matrix $A$ have the form

$$
\lambda_{2i-1,2i} = -\frac{e + \mu k \pm R}{2d}, R_i = \sqrt{4\mu k + (e - \mu k)^2}, i \in I
$$

(10)

$$
\begin{bmatrix}
\frac{1}{d} - \lambda_i I_n & (-1/d)I_n \\
(-1/dk)L & (-e/dk - \lambda_i)I_n
\end{bmatrix}
\begin{bmatrix}
v_i,\theta \\
v_{i,p}
\end{bmatrix} = 0_{2n}
$$

(13)

where each eigenvector in (11) and (12) is partitioned into two $n \times 1$ vectors according to the structure of $A$. Then, from the first $n$ rows of (13), $v_{i,p}$ can be written as

$$
v_{i,p} = -(L + d\lambda_i I_n)v_{i,\theta},
$$

(14)

and hence, from the second group of $n$ rows in (13) we obtain

$$
(L + (e + dk\lambda_i)(L + d\lambda_i I_n))v_{i,\theta} = 0_n.
$$

(15)

We consider the eigenvalue-eigenvector pairs proposed in the statement of the theorem. The proof proceeds by first verifying that $(\lambda_i, v_{i,\theta})$ satisfy (15). Then, the affirmed pair is replaced in (14) to confirm the proposed expression for $v_{i,p}$.

We first investigate the eigenstructure related to the first two eigenvalues where substituting $\mu_1 = 0$ [see Lemma 2.2(b)] into (10) yields $\lambda_1 = 0$ and $\lambda_2 = -e/(dk)$.

Associated with $\lambda_1 = 0$ is the eigenvector $v_1 = [v_{1,\theta}^T, v_{1,p}^T]^T = [u_1^T, 0_n^T]^T$. Replacing $(\lambda_1, v_{1,\theta})$ in (15) yields

$$
(L + Le)u_1 = (e - 1)Lu_1 = 0_n,
$$

where the above is obtained on account of $Lu_1 = 0_1 u_1 = 0_n$. Then, from (14) we have $v_{1,p} = -Lu_1 = 0_n$ which confirms the validity of $(\lambda_1, v_1)$ as an eigenvalue-eigenvector pair of the matrix $A$.

For the second eigenvalue of the matrix $A$, from (11), corresponding to $\lambda_2 = -e/(dk)$ we have $v_2 = [v_{2,\theta}^T, v_{2,p}^T]^T = [(k/e)u_1^T, u_1^T]^T$. Substituting $(\lambda_2, v_{2,\theta})$ into (15) results in

$$
\begin{bmatrix}
0 \\
-L + (e + dk(-e/dk))(L + d(-e/dk)I_n)
\end{bmatrix}u_1(k/e) = 0_n
$$

where we used $Lu_1 = \mu_1 u_1 = 0$. From (14), $v_{2,p}$ is then

$$
v_{2,p} = -(L + d(-e/dk)I_n)u_1(k/e) = -Lu_1(k/e) + (e/k)u_1(k/e) = u_1
$$

which together with $v_{2,\theta}$ show the validity of $(\lambda_2, v_2)$.

Next, we show that for the remaining eigenvalues, the eigenvectors proposed in (12) satisfy (14) and (15). For simplicity we drop the subindex $i$ and write

$$
\lambda = \frac{e + \mu k \pm R}{2dk}, R = \sqrt{4\mu k + (e - \mu k)^2}
$$

(16)

$$
v = \left[\begin{array}{c}
u(e + dk\lambda)/\mu \\
-u
\end{array}\right] = \left[\begin{array}{c}
v_{0} \\
v_{p}
\end{array}\right].
$$

(17)

Substituting the pair $(\lambda, v_0)$ into (15) and disregarding the coefficient of $u$ in $v_0$ as it has no impact on the zero result lead to

$$
(L + (e + dk\lambda)(L + d\lambda I_n))u = -Lu + (e + dk\lambda)(Lu + d\lambda u)
$$

$$
= -\mu u + (e + dk\lambda)(\mu u + d\lambda u)
$$

$$
= -\mu u + (e + \mu k \pm R)(e + \mu k \pm R)2k
$$

$$
= -\mu u + \frac{1}{4k}(R^2 - (e - \mu k)^2)u
$$

$$
= -\mu u + \frac{1}{4k}(4\mu k + (e - \mu k)^2) - (e - \mu k)^2)u
$$

$$
= -\mu u + \frac{1}{4k}(4\mu k)u = 0_n.
$$

(18)

Then, (14) is proven as follows:

$$
v_p = -(L + d\lambda I_n)v_0
$$

$$
= -(Lu + d\lambda u)(e + dk\lambda)/\mu
$$

$$
= -(\mu u + d\lambda u)(e + dk\lambda)/\mu
$$

$$
= -(\mu + d\lambda)(e + dk\lambda)u/\mu = -u.
$$

(19)

From (18) and (19), it is clear that $(\lambda, v)$ in (16), (17) is an eigenvalue-eigenvector pair of the matrix $A$. ■

**Theorem 2.3** derived expressions for the eigenvalues and eigenvectors of the matrix $A$ in the microgrid model (7)–(8).

Through the obtained eigenstructure of the microgrid, one can exploit a change into modal coordinates to investigate the system stability properties. Define the associated matrices

$$
A = \text{diag}(\lambda_1, \ldots, \lambda_{2n}), \quad V = [v_1 \ldots v_{2n}].
$$
We then consider the state transformation $x = Vz$. From (7)–(9) and noting that $\Lambda = V^{-1}AV$, the transformed state $z$ satisfies
\[
\dot{z} = \Lambda z + V^{-1}Hf + V^{-1}\bar{P}
\] (20)
where, by direct computation,
\[
V^{-1}H = \Gamma U_H, \quad V^{-1}\bar{P} = -\Gamma U_P,
\] (21)
with
\[
\Gamma = \text{diag}\{ e^{-1}, -\lambda_2, \lambda_3, -\lambda_4, \ldots, -\lambda_{2n-1}, -\lambda_{2n} \},
\]
\[
U_H = u_h \odot \left[ \begin{array}{c} 1 \\ \vdots \\ 1 \end{array} \right], \quad U_P = u_p \odot \left[ \begin{array}{c} 1 \\ \vdots \\ 1 \end{array} \right]
\]
(22)
\[
u_h = R^{-1}U^{-1}BY, \quad u_p = R^{-1}U^{-1}P^*
\]
(23)
where $u_h \in \mathbb{R}^{n \times m}$, $u_p \in \mathbb{R}^{n \times 1}$ and $R = \text{diag}\{R_i\}_{i \in \mathcal{I}}$. We will show in the following section that the transformed state $z$ satisfies (20)–(24) has a special structure convenient for stability analysis.

III. STABILITY ANALYSIS

The closed-loop system (20) can be regarded as a linear system with a nonlinear ‘perturbation’ term (the second term) affected by bounded disturbances (the third term). Under certain conditions on the nonlinear term one can expect the linear part of the dynamics to dominate and, if the latter is stable, ultimately boundedness of the trajectories starting inside a region of the state space may be achieved [10]. In this regard, in this section we start by addressing the stability of the linear part of system (20) and follow progressively steps to finally establish the ultimate boundedness of the trajectories of the full nonlinear system, thus providing stability conditions that go beyond local stability around the equilibrium point. It is worth noting that another analysis that considers a model including nonlinearities in power systems has been presented in [18].

A. Stability of the System’s Linear Part

To begin with, the stability of the linear part of system (20) is established by analysing its eigenvalues.

Lemma 3.1: The matrix $A$ in (7) (equivalently, $\Lambda$ in (20)) has stable (real negative) eigenvalues, except for $\lambda_1 = 0$ which represents the rotational symmetry of the system.

Proof: As can be seen in (10), the eigenvalues of the matrix $A$ are functions of the eigenvalues of the Laplacian matrix $L$. It is well-known that the Laplacian matrix is a positive semi-definite matrix and hence, its eigenvalues $\mu_i$ are positive except for the zero eigenvalue $\mu_1 = 0$ representing the rotational symmetry.

Each eigenvalue $\lambda_i$, $i \neq 1$ is stable if and only if
\[
\frac{e + \mu_i k}{2d} = \sqrt{\frac{4\mu_i k + (e - \mu_i k)^2}{2dk}} < 0 \iff \sqrt{\frac{4\mu_i k + (e - \mu_i k)^2}{2dk}} < e + \mu_i k \iff 4\mu_i k + (e - \mu_i k)^2 < (e + \mu_i k)^2 \iff 4\mu_i k < 4\mu_i k e \iff 1 < e,
\]
which is always true since $e = 1 + ed > 1$ for $e, d > 0$. Therefore, apart from the zero eigenvalue $\lambda_1 = 0$, prevalent to systems with the Laplacian matrix representation, the eigenvalues of the matrix $A$ are real negative numbers, thus stable.

B. Model Decoupling Property

In view of facilitating the stability analysis, the structure of the closed-loop system (20)–(24) can be unfolded one step further by using a property of the eigenvector matrix of the Laplacian $L$, as per the following remark.

Remark 3.2: According to Lemma 2.2(b), the first rows of the matrices $u_h$ and $u_p$ in (24)–(24) are, respectively,
\item $u_h(1,:) = [U^{-1}(1,:)BY/R_1 = 0^T_m,$
\item $u_p(1,:) = [U^{-1}(1,:)], P^*/R_1 = (\sum_{i=1}^n P_i^*)/ne,$
where we have also used the structure of the incidence matrix $B$ and $R_1 = e$ (see (10) for $\mu_1 = 0$).

Letting $z = [z_1 \ z_2 \ z^T]^T$, $\dot{z} = [\dot{z}_1 \ \dot{z}_2 \ \dot{z}^{p, \ell} T]^T$, using (22) with $\mu_2 = -e/(dk)$, and exploiting Remark 5.2 we have
\[
\dot{z} = \Lambda \dot{z} + \hat{\Gamma} \dot{U}_H f - \dot{U}_P
\]
(26)
where
\[
\hat{\Lambda} = \text{diag}(\lambda_3, \lambda_4, \ldots, \lambda_{2n}),
\]
\[
\hat{\Gamma} = \text{diag}(\lambda_3, -\lambda_4, \ldots, -\lambda_{2n-1}, -\lambda_{2n}),
\]
\[
\dot{U}_H = [U_H]_{(3:2n,:), \ldots}, \quad \dot{U}_P = [U_P]_{(3:2n,:)}.
\]

In the next step, the two subsystems (25) and (26) are shown to be decoupled from each other. To this purpose, we study the dependency of the function $f$ on the $z$ states.

Lemma 3.3: The system (26) consisting of the last $2n - 2$ states is decoupled from the system (25).

Proof: From $x = Vz$, if the matrix $V$ with columns given by (11)–(12) is partitioned as
\[
V = \begin{bmatrix} V_0 \\
V_p \end{bmatrix} = \begin{bmatrix} u_1 \\
e^{(k/e)u_1} \ldots \end{bmatrix},
\]
(30)
where $V_0, V_p \in \mathbb{R}^{n \times 2n}$, yields $x_0 = V_0 z$. Then we have, using the structure of the incidence matrix $B$,
\[
[\theta_i - \theta_j]_{i,j \in \mathcal{E}} = B^T V_0 B.
\]
(31)
Using $u_1 = 1_n$ (see Lemma 2.2(b)) and the fact that the matrix $B^T$ has just two nonzero elements $\{ -1, 1 \}$ in each of its rows, the first two columns of the matrix $B^T V_0$ are always zero and hence, (31) does not depend on $(z_1, z_2)$.

That is, $\lambda_i = [\theta_i - \theta_j]_{i,j \in \mathcal{E}}$ does not depend on $(z_1, z_2)$ and thus, system (26) is decoupled from system (25).

From Lemma 3.1 the linear subsystem (25) has one zero and one stable eigenvalue. According to Lemma 3.1 and Lemma 3.3 the companion subsystem (26) has a stable diagonal linear part and a nonlinear perturbation term that depends only on its own state variables. In the following two sections we study the boundedness properties of these decoupled subsystems.
C. Boundedness of the Average Frequency Error

The representation (25)–(26) of the microgrid system facilitates the analysis of the average frequency error and its boundedness, as shown below.

Lemma 3.4: For the microgrid system represented by (25)–(26), the average frequency error \( \omega_{\text{sync}} \) given in (6) is bounded if all the inverter power injection errors \( P_i^*, i \in I \), are bounded.

Proof: From \( x = V z \) it can be shown that
\[
\dot{z}_1 = (\sum_{i=1}^{n} \frac{\theta_i}{n}) - k \dot{z}_2 / e. \tag{32}
\]
Since from (25), \( \dot{z}_2 = \lambda_2 z_2 + (\sum_{i=1}^{n} P_i^*)/(ndk) \) with \( \lambda_2 = -\epsilon/(dk) \), then \( \dot{z}_2 \) and \( z_2 \) remain bounded for bounded \( P_i^*, i \in I \). Furthermore, from (25) and (32), we have
\[
\dot{z}_1 = (\sum_{i=1}^{n} \frac{\theta_i}{n}) - k \dot{z}_2 / e = (e - 1) (\sum_{i=1}^{n} P_i^*)/(nde). \tag{33}
\]
Then the average frequency error \( \omega_{\text{sync}} \) also remains bounded for bounded inverter errors \( P_i^*, i \in I \).

Corollary 3.5: The average frequency error \( \omega_{\text{sync}} \) converges to
\[
\omega_{\text{sync}} \equiv (\sum_{i=1}^{n} P_i^*)/(nd) \tag{34}
\]
if \( \sum_{i=1}^{n} P_i^* \) is constant.

Proof: From the \( z_2 \) equation in (25), \( z_2 \) is proved to converge to a constant and hence \( \dot{z}_2 \) converges to zero. Then, from (33) and \( e = 1 + ed \), \( \omega_{\text{sync}} \) converges to (34).

Remark 3.6: From Corollary 3.5, a smaller value of \( \epsilon \) yields a smaller average frequency steady state error.

D. Ultimate Boundedness

We will analyze the ultimate boundedness properties of the subsystem (26) by applying Theorem 3 of [9]. When specialised to non-switched systems, the latter result establishes that for a stable linear system with a nonlinear perturbation term, the trajectories starting inside a region of the state space are ultimately bounded if the nonlinear perturbation satisfies certain conditions. More specifically, to meet the requirements of [9, Theorem 3], the perturbation term should be bounded by a componentwise non-increasing (CNI) function and further satisfy a contractivity condition. We first derive in the following result a CNI bound for the perturbation term in (26) and then address the contractivity condition in Lemma 3.8.

Lemma 3.7: The perturbation term \( \hat{T}(\hat{U}_H f - \hat{U}_P) \) in system (26) is bounded by a CNI function as follows:
\[
|\hat{T}(\hat{U}_H f - \hat{U}_P)| \leq |\hat{T}_H| F(\hat{z}) + |\hat{T}_P|, \tag{35}
\]
where
\[
F(\hat{z}) \equiv \frac{([B^T V_\theta ||z||]_3)}{6}, \tag{36}
\]
with \( V_\theta \) as in (30).

Proof: We first bound the nonlinear function \( f \), with components \( f(\theta_i - \theta_j) \) with \( f(x) = \sin(x) - x \). Recalling from the proof of Lemma 3.3 that \( \| \theta \| \) only depends on \( \dot{z} \), and using the inequality \( |\sin(x) - x| \leq |x|^3/6 \) we can bound
\[
f(\theta_i - \theta_j) \leq \frac{([B^T V_\theta ||z||]_3)}{6} = F(\hat{z}), \tag{37}
\]
yielding
\[
f = f(\theta_i - \theta_j)]_{i,j} \leq \frac{([B^T V_\theta ||z||]_3)}{6} = F(\hat{z}). \tag{38}
\]
The bound (35) then follows. The CNI property of the bound is immediate from the nonnegativity of all entries in the products involved.

Following [9], we next define a nonlinear mapping \( T : \mathbb{R}^{2n-2} \rightarrow \mathbb{R}^{2n-2} \) constructed from the bound (35) as follows:
\[
T(\hat{z}) = |\hat{A}|^{-1} (|\hat{T}_H| F(\hat{z}) + |\hat{T}_P|) = |\hat{U}_H| F(\hat{z}) + |\hat{U}_P|, \tag{39}
\]
where the second line follows from (27) and (28). From [9, Theorem 3] (see [8] for proofs), if a vector \( \hat{z} \) with positive components exists such that contractivity condition
\[
T(\hat{z}) < \hat{z} \tag{40}
\]
holds componentwise, then the trajectories of the nonlinear system (26) are ultimately bounded and the ultimate bound can be found by recursively iterating the mapping \( T(\cdot) \) starting from \( \hat{z} \). In the following lemma we give a sufficient condition for (40) to hold for some \( \hat{z} \).

Lemma 3.8: Suppose there exist positive constants \( g_1, g_2, \ldots, g_{n-1} \) such that the scalar inequality
\[
u_p^2 < \frac{4g_1^3}{27\gamma_i} \tag{41}
\]
holds for \( i = 1, \ldots, n-1 \), where \( u_p \) is defined in (24), \( \gamma_i = |u_{h(i+1, i)}| ([B^T V_\theta] G)^{3/2} > 0 \), with \( u_h \) defined in (24) and \( G \equiv [1, 1, g_1, g_1, g_2, g_2, \ldots, g_{n-1}]^T \). Then there exists a scalar \( \zeta > 0 \) such that the nonnegative vector \( \hat{z} \) satisfies the contractivity condition (40).

Proof: From (27), it can be seen that each even row of \( U_H \) and \( U_P \) is equal to its preceding row and thus, \( \hat{U}_H, \hat{U}_P \) defined in (29) and the vector function \( T(\cdot) \) in (39) also share the same property. That is, letting \( T(\hat{z}) = [T_{2i-1}(\hat{z}) \ T_{2i}(\hat{z}) \ldots T_{2n-2}(\hat{z})] \), we have for \( i = 1, \ldots, n-1 \)
\[
T_{2i-1}((\hat{z})) = t_i((\hat{z}) \otimes [1]), \tag{42}
\]
\[
t_i((\hat{z}) = |u_{h(i+1, i)}| F(\hat{z}) + |u_{p(i+1)}| \tag{43}
\]
where
\[
T_{2i-1}(\hat{z}) T_{2i}(\hat{z}) = t_i((\hat{z}) \otimes [1] < \frac{\hat{z}_{2i-1}}{\hat{z}_{2i}}, \tag{44}
\]
with \( V_\theta \) as in (30).
which, by choosing \( \bar{\varepsilon} \) to have pairwise repeated rows, can be further simplified to
\[
\begin{bmatrix}
T_{2i-1}(\bar{\varepsilon}) \\
T_{2i}(\bar{\varepsilon})
\end{bmatrix} = t_i(\bar{\varepsilon}) \otimes \begin{bmatrix}1 \\
1
\end{bmatrix} < \bar{\varepsilon}_{2i} \otimes \begin{bmatrix}1 \\
1
\end{bmatrix},
\]
and hence,
\[
t_i(\bar{\varepsilon}) = |u_{h(i+1)}| \left(\frac{|B^T B|}{6} \right) + |u_{p(i+1)}| < \bar{\varepsilon}_{2i},
\tag{44}
\]
for \( i = 1, \ldots, n - 1 \). Further substituting \( z = G\zeta \) and \( \bar{\varepsilon} = G_{(3:2n)} \zeta \) with \( G = [1, 1, g_1, g_2, \ldots, g_{n-1}]^T \), yields
\[
\ell_i(\zeta) = |u_{h(i+1)}| \left(\frac{|B^T B|}{6} \right) \zeta^3 + |u_{p(i+1)}| < g_i \zeta \tag{45}
\]
for \( i = 1, \ldots, n - 1 \), where \( \ell_i(\zeta) = t_i(\bar{\varepsilon}) \). Equivalently,
\[
\gamma_i \zeta^3 - g_i \zeta + |u_{p(i+1)}| < 0
\tag{46}
\]
where \( \gamma_i = |u_{h(i+1)}| \left(\frac{|B^T B|}{6} \right) > 0 \).

For a generic cubic function \( Q(\zeta) = a\zeta^3 + b\zeta^2 + c\zeta + d \), it is known that to have \( Q(\zeta) < 0 \) for \( \zeta > 0 \), \( Q(\zeta) \) must have three distinct real roots, which is guaranteed if its discriminant \( \Delta = 18abcd - 4b^3d + b^2c^2 - 4ac^3 - 27a^2d^2 \) is positive. For the cubic function on the left hand side of \( 46 \) the positive discriminant condition takes the form
\[
\Delta_i = \gamma_i(4\gamma_i^3 - 27\gamma_i^2p^2) > 0,
\]
which coincides with \( 41 \).

We observe that the contraction condition \( 41 \) can be loosely interpreted as a tolerance on ‘how dissimilar’ the inverter power errors, \( P^*_i \), are allowed to be to meet the desired requirements. Indeed, from Lemma 2.2.3 and the definition of \( u_p \) in \( 24 \), for \( i = 1 : n - 1 \),
\[
u_{i+1} = \ell_i(\gamma^*_{i+1}),
\tag{47}
\]
is a linear combination of the inverter power errors such that, if \( P^*_i = P^*_j \) for all \( i, j \in I \) we have \( u_p(i + 1) = 0 \) for \( i = 1 : n - 1 \) and condition \( 41 \) is automatically satisfied.

We now have all the elements to establish the stability properties of the droop controlled microgrid system.

**Theorem 3.9:** Under the conditions of Lemma 3.3 let \( \zeta > 0 \) satisfy \( 46 \). Then, for the microgrid system represented by \( 25 \)–\( 26 \), the average frequency error \( \omega_{sync} \) given in \( 6 \) is bounded and the trajectories of subsystem \( 26 \) with initial conditions satisfying \( |\bar{\varepsilon}(0)| \leq G\zeta \) are ultimately bounded as \( \lim_{t \to \infty} |\bar{\varepsilon}(t)| \leq \lim_{t \to \infty} T^k(G\zeta) \).

**Proof:** Immediate from the results in this section and Theorem 3 of [9].

### IV. Example

To illustrate the discussed concepts, we consider an academic example of a microgrid system consisting of three inverter buses and two edges with \( a_{12} = 2 \), \( a_{13} = 5 \), \( a_{23} = 0 \). For this system, the graph data, the incidence matrix, the Laplacian matrix and its eigenstructure, after removing all zero rows and columns corresponding to \( a_{23} = 0 \), are given by
\[
B = \begin{bmatrix}
1 & 0 & 0 \\
0 & 1 & 0 \\
1 & -2 & -5
\end{bmatrix}, \quad Y = \begin{bmatrix}2 & 0 \\
0 & 1
\end{bmatrix}, \quad L = \begin{bmatrix}1 & 0 & -1.2718 \\
0 & 1 & 0.2718
\end{bmatrix},
\]
\[
M = \text{diag}\{0, 2.6411, 11.5389 \}, \quad U = \begin{bmatrix}0.4718 & -1.2718 \\
-1.2718 & 0.4718
\end{bmatrix}.
\]

Using the above data, the system matrices and its eigenvalue-eigenvectors from \( 8 \), \( 10 \), \( 12 \), with \( d = 1 \), \( k = 1 \) and \( \epsilon = 1 \) giving \( \epsilon = 1 + \epsilon d = 2 \), are
\[
A = \begin{bmatrix}
-7 & 2 & 5 & -1 & 0 & 0 \\
2 & -2 & 0 & -1 & 0 & 0 \\
5 & 0 & -5 & 0 & 0 & -1 \\
-7 & 2 & 5 & -2 & 0 & 0 \\
2 & -2 & 0 & -2 & 0 & 0 \\
5 & 0 & -5 & 0 & -2 & 0
\end{bmatrix}, \quad H = \begin{bmatrix}2 & -5 \\
-5 & 2
\end{bmatrix},
\]
\[
\Lambda = \text{diag}\{0, -2, -0.6641, -3.9770, -0.9126, -12.4463 \},
\]
\[
V = \begin{bmatrix}
0.5 & 0.2386 & -0.3532 & -0.1217 & 1.1696 \\
1.05 & -0.7444 & 1.1017 & 0.0260 & -0.2499 \\
1.05 & 0.5058 & -0.7486 & 0.0957 & -0.9197 \\
0 & 1 & -0.4718 & -0.4718 & 1.2718 \\
0 & 1 & 1.4718 & -0.2718 & -0.2718 \\
0 & 1 & -1 & -1 & -1
\end{bmatrix}.
\]

Next, to form the transformed system \( 20 \) with matrices \( 21 \), the required matrices \( 22 \)–\( 25 \) are
\[
\Gamma = \text{diag}\{1.2, -0.6641, 3.9770, -0.9126, 12.4463 \},
\]
\[
R = \text{diag}\{2.3.3129, 11.5336 \},
\]
\[
u_h = R^{-1}U^{-1}BY = R^{-1}\begin{bmatrix}
1 & 0.4718 & -1.2718 \\
-1.2718 & 0.4718 & 1
\end{bmatrix}^{-1} \begin{bmatrix}1 & 1 \\
0 & -1
\end{bmatrix}Y = \begin{bmatrix}0.3462 & -0.2353 \\
-0.2353 & 0.3462
\end{bmatrix},
\]
\[
u_p = R^{-1}U^{-1}P^* = R^{-1}\begin{bmatrix}
1 & 0.4718 & -1.2718 \\
-1.2718 & 0.4718 & 1
\end{bmatrix}^{-1} \begin{bmatrix}P^*_1 & P^*_2 \\
P^*_2 & P^*_3
\end{bmatrix} = \begin{bmatrix}0.3333 & 0.3333 & 0.3333 \\
0.1392 & -0.4343 & 0.2951
\end{bmatrix},
\]
\[
\Lambda = \text{diag}\{0.1667, 0.1667, 0.1667, 0.0420, -0.1311, 0.0891 \\
-0.0410, 0.0088, 0.0322
\},
\]
\[
(P^*_1 + P^*_2 + P^*_3)/6 = \begin{bmatrix}P^*_1 & P^*_2 & P^*_3
\end{bmatrix}
\]
\[
V^{-1}H = \Gamma U_H = \begin{bmatrix}0 & 0 & 0 \\
0 & 0 & 0 \\
-0.2299 & 0.1562 & 1.3776 \\
-0.2299 & 0.1562 & 1.3776 \\
0.0909 & 0.3340 & -1.2378 & -4.5456
\end{bmatrix},
\]
\[
V^{-1}P = -\Gamma U_P = \begin{bmatrix}
(P^*_1 + P^*_2 + P^*_3)/6 \\
(P^*_1 + P^*_2 + P^*_3)/3
\end{bmatrix}.
\]

It can be seen that \( u_{h(1,:)} = 0^T_2 \) yields the first two rows of
$V^{-1}H$ equal to zero, which confirms that the two subsystems

$$
\begin{bmatrix}
\dot{z}_1 \\
\dot{z}_2 \\
\dot{z}_3 \\
\dot{z}_4
\end{bmatrix} = \begin{bmatrix} 0 & 0 & 1/2 & 0 \\ 0 & 0 & 0 & 0 \\ -0.6641 & 0 & 0 & -3.9770 \\ 0 & -0.9126 & 0 & -12.4463 \\
\end{bmatrix} \begin{bmatrix}
\dot{z}_1 \\
\dot{z}_2 \\
\dot{z}_3 \\
\dot{z}_4
\end{bmatrix} + \begin{bmatrix}
\ell_1(\dot{z}_1, \dot{z}_2, \dot{z}_3, \dot{z}_4)
\end{bmatrix} / 3
$$

are decoupled.

Partitions $V$ as in (30) with $V_\theta = V_{(1:3:j)}$, the line phases are computed from (31) to be

$$
\begin{bmatrix}
\theta_1 - \theta_4 \\
\theta_2 - \theta_3 \\
\theta_3 - \theta_4
\end{bmatrix} = B^TV_\theta z
$$

$$
= \begin{bmatrix} 0 & 0.9831 & -1.4549 & -0.1478 & 1.4195 & -0.2672 & 0.3954 & -0.2175 & 2.0893 \\ 0 & -0.2672 & 0.3954 & -0.2175 & 2.0893 & 0.9831 & -1.4549 & -0.1478 & 1.4195 \\ 0 & 0.3954 & -0.2175 & 2.0893 & -0.2672 & 0.3954 & -1.4549 & -0.1478 & 1.4195 \\
\end{bmatrix}
$$

The function $F(\hat{z})$ in Lemma 3.1 is obtained from (36) as

$$
F(\hat{z}) = \left(\frac{||B^TV_\theta|||z||^3}{6}\right) / 6
$$

$$
= \left(\begin{bmatrix} 0.9831 & -1.4549 & -0.1478 & 1.4195 & -0.2672 & 0.3954 & -0.2175 & 2.0893 \\ -0.2672 & 0.3954 & -0.2175 & 2.0893 & 0.9831 & -1.4549 & -0.1478 & 1.4195 \\ 0.3954 & -0.2175 & 2.0893 & -0.2672 & 0.3954 & -1.4549 & -0.1478 & 1.4195 \\
\end{bmatrix}\right)^3
$$

Then, from (39), (42)–(43) the nonlinear mapping $T : \mathbb{R}_+^4 \rightarrow \mathbb{R}_+^4$ is

$$
T(\hat{z}) = [\hat{U}_H(F(\hat{z}) + [\hat{U}_P] = [\ell_1(\hat{z})] \otimes [1
$$

with

$$
t_1(\hat{z}) = |u_{h1}(\hat{z})| |F(\hat{z}) + [\hat{U}_P]| = 0.3462 - 0.3535 + 0.0420P*_{1} - 0.1311P*_{2} + 0.0891P*_{4},
$$

$$
t_2(\hat{z}) = |u_{h2}(\hat{z})| |F(\hat{z}) + [\hat{U}_P]| = -0.0995 - 0.3659 + 0.0410P*_{1} - 0.0008P*_{2} + 0.0322P*_{4},
$$

With the selection of $z_3 = z_4 = g_1\zeta$ and $z_5 = z_6 = g_2\zeta$, the scalar inequalities (45) to satisfy the contractivity condition (40) are

$$
t_1(\zeta) = \left(0.3462(0.9831z_1 - 1.4549z_2 + 0.1478z_3 + 1.4195z_4) + 0.0420P*_{1} - 0.1311P*_{2} + 0.0891P*_{4}\right) = g_1\zeta^3
$$

$$
t_2(\zeta) = \left(0.0995(0.9831z_1 - 1.4549z_2 + 0.1478z_3 + 1.4195z_4) + 0.0410P*_{1} - 0.0008P*_{2} + 0.0322P*_{4}\right) = g_2\zeta^3
$$

for arbitrary $g_1, g_2 > 0$. The inverter power injection set-points, through the linear functions (47), then need to satisfy the scalar inequalities

$$
u_{P(2)}^2 = \left(\ell_1(P^*_1, P^*_2, P^*_3)\right)^2 < \frac{4g_1^3}{27\gamma_1} \Rightarrow b_1,
$$

$$
u_{P(3)}^2 = \left(\ell_2(P^*_1, P^*_2, P^*_3)\right)^2 < \frac{4g_2^3}{27\gamma_2} \Rightarrow b_2
$$

for the system to be ultimately bounded. With regard to these inequalities, one can run a nonlinear optimisation on $g_1$ and $g_2$ to maximise the upper bounds $b_1$ and $b_2$. The nonlinear optimisation

$$
\max \min \{b_1, b_2\}
$$

yields $g_1 = 8.0377, g_2 = 6.4202$ which in turn lead to $b_1 = 0.0421, b_2 = 0.0421$.

Take, for instance, $P^*_1 = 1, P^*_2 = 2, P^*_3 = 3$. The contractivity conditions (48)–(49) are then satisfied

$$
u_{P(2)}^2 = 0.0471^2 < 0.0421 = b_1,
$$

$$
u_{P(3)}^2 = 0.0732^2 < 0.0421 = b_2.
$$

The next step is to find $\hat{z} = G\zeta$. For each $t_i(\hat{z})$ function, the $\zeta$ domain for which $t_i(\hat{z}) < g_i\zeta$ is the interval between the two positive roots of the polynomial $Q_i(\zeta) = t_i(\hat{z}) - g_i\zeta = 0$. For $i = 1, 2$ we have roots($Q_1$) = $\{-0.0665, 0.0003, 0.0662\}$ and roots($Q_2$) = $\{-0.0835, 0.0008, 0.0827\}$ which yields

$$
\zeta(Q_1) = (0.0003, 0.0662), \quad \zeta(Q_2) = (0.0008, 0.0827).
$$

Then, the $\zeta$ domain that satisfies both conditions is the intersection of these intervals, that is,

$$
\zeta \in \zeta(Q_1) \cap \zeta(Q_2) = (0.0008, 0.0662).
$$

Now we just need to select a starting point $\zeta_0$ from this interval, compute the associated $\hat{z}_0$ and iteratively calculate the ultimate bound of the system. From (9), the ultimate bound can be computed by first taking $\hat{z}_0 = G(\zeta, 2\zeta)\zeta_0$, $T(\hat{z}_1) = T(\hat{z}_0)$ and then iterating, $T(k+1)(\hat{z}) = T(k)(\hat{z})$ for $k \in \mathbb{Z}_+$. Since $T(k+1)(\hat{z}) \leq T(k)(\hat{z})$, the ultimate bound is obtained as $\lim_{k \rightarrow \infty} T(k)(\hat{z}) = b_2 > 0$.

Let $\zeta_0 = 0.0327$. Then $\hat{z}_0 = (0.2628, 0.2628, 0.2099, 0.2099)^T$. The resulting ultimate bound on the $\hat{z}$ states is

$$
b_2 = \begin{bmatrix}
0.0481 \\
0.0481 \\
0.0739 \\
0.0739
\end{bmatrix}.
$$

We can interpret this ultimate bound on the line phases $[\theta_1 - \theta_4]_{i,j \in J}$ as follows

$$
\left|\frac{\theta_1 - \theta_4}{\theta_1 - \theta_4}\right| \leq |B^TV_\theta||z|| \leq |B^TV_\theta| / |b_2|
$$

where the * entries are irrelevant since the first two columns of $B^TV_\theta$ are zero. The above bounds on the phase differences is validated as can be seen in Fig. (1.a).

The next variable derived from this simulation is the average frequency error as in (3). Corollary 3.3 proves that this frequency converges to the steady state frequency $\omega_{\text{sync}}$ as in (13). It is also notable that this steady state average frequency static error is reliant on $\epsilon$, that is, by
decreasing $\epsilon$ we obtain a smaller $\omega_{\text{sync}}$. For $\epsilon = 1$ and $\epsilon = 0.1$ the obtained values are

$$
\epsilon = 1 \rightarrow \omega_{\text{sync}} = 1, \quad \epsilon = 0.1 \rightarrow \omega_{\text{sync}} = 0.1818.
$$

The convergency of $\omega_{\text{sync}}$ to $\omega_{\text{sync}}$ for $\epsilon = 1$ is depicted in Fig. 1(b).

![Fig. 1. (a) Line phases, (b) convergence of $\omega_{\text{sync}}$ to $\omega_{\text{sync}}$](image)

V. CONCLUSIONS

We have analysed theoretical properties of inverter-based microgrids controlled via primary and secondary loops. We have shown that frequency regulation is ensured without the need for time separation, and that ultimate boundedness of the trajectories starting inside a region of the state space is guaranteed under a condition on the inverters power injection errors. The trajectory ultimate bound can be computed by simple iterations of a nonlinear mapping and provides a certificate of the overall performance of the controlled microgrid. Future work includes the derivation of design procedures based on the provided analysis, the extension of the results to more general controller parameters and structures as well as relaxing some of the modelling assumptions.

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